

Durham E-Theses

Qrgano – Metallic: Co-ordination compounds of group IB

Parkin, C

How to cite:

Parkin, C (1962) Qrgano – Metallic: Co-ordination compounds of group IB, Durham theses, Durham University. Available at Durham E-Theses Online: http://etheses.dur.ac.uk/9246/

Use policy

The full-text may be used and/or reproduced, and given to third parties in any format or medium, without prior permission or charge, for personal research or study, educational, or not-for-profit purposes provided that:

- a full bibliographic reference is made to the original source
- a link is made to the metadata record in Durham E-Theses
- the full-text is not changed in any way

The full-text must not be sold in any format or medium without the formal permission of the copyright holders.

Please consult the full Durham E-Theses policy for further details.

ORGANO - METALLIC

CO-ORDINATION COMPOUNDS OF GROUP IB

by

C. PARKIN B.Sc.

A thesis submitted for the Degree of Doctor of Philosophy in the University of Durham.



PREFACE

The work described in this thesis is the original work of the author except where acknowledged by reference, has not been submitted for any other degree and has been carried out under the supervision of Professor G. E. Coates, at the Science Laboratories of the Durham Colleges in the University of Durham, between September 1959 and June 1962. It is also the subject of two papers, published with Professor Coates, entitled: (a) Ethynylcopper(I) Complexes with Tertiary Phosphines, <u>J. Inorg. Nuclear Chem</u>, 1961, <u>22</u>, 59, (b) Gold(I) Ethynyls and their Co-ordination Complexes, <u>J.</u>, in the press.

The author wishes to express his sincere thanks to Professor G. E. Coates for his invaluable advice and encouragement throughout the work. He is grateful to Dr. H.M.M. Shearer, and Mr. P.W.R. Corfield for their interest in the crystal structures of some of the compounds described and to Dr. F. Glockling for helpful discussions. He is also indebted to the Ethyl Corporation for a maintenance grant.

June 1962

(i)

CONTENTS

Preface.	ø		٠	0	•		•	•	6		•	•	•	•	. 0	•	e :	. 0		• •	. ((i)	
Contents	•	e	•	e	•		9	•	•	•	•		•	•	•	•	•	•	۰.	•	. ((i	i))
Abstract	•	•	•	•	•	٠	•	•	•	•	•	•	•	•	•	•	•	•	•	• •	,	(ii)

INTRODUCTION

Co-ordination Chemistry of Copper(I), Silver(I) and Gold(I)

complexes with halogens nitrogen oxygen and sulphur	4
phosphorus and arsenic complexes	5
cyanide complexes	8
isocyanide complexes	9
carbon monoxide complexes	10
alkyls and aryls	12
ethylene complexes	15

Transition metal ethynyl complexes

"olefine- type" complexes with disubstituted acetylenes	17
complexes with an acetylene acting as a bidentate ligand	18
complexes with σ -bonding between the metal and the acetylene	19
copper and silver ethynyl complexes	21
goldethynyl complexes	25

EXPERIMENTAL

CHAPTER I Preparation and Purification of reagents

Attempted preparation of trialkylphosphines	
from trialkylaluminiums	26
Preparation of trimethylphosphine	28
Preparation of triethylphosphite	30
Preparation of triethylstibine	30
Preparation of prop-l-yne (methylacetylene)	30
Preparation of 3,3;dimethyl-l-butyne (t-butylacetylene)	32
Preparation of <u>p</u> -bromophenylacetylene	33
Preparation of mesitylacetylene	35
Preparation of 1-phenylbutadiyne	37
Preparation of copper(I) chloride	39
Purification of solvents	40

CHAPTER II Experimental Methods

Α.	Molecular weight determinations	
	cryoscopically	43
	ebullioscopically	44
в.	Dipole Moment measurements	45
C.	Infra-Red Spectra	59
D.	Quantitative Analyses	60

CHAPTER III Copper, Silver and Mercury Acetylides

Preparation and solubility properties	s of 63
Attempted preparations of "crystallin	1e#
phenylethynylcopper	65

	Preparation and solubility properties of silver acetylides	66
	Alkyl and aryl ethynylmercury(II) compounds	67
diam'ny artes P		an an an Arrange An Arrange an Arrange Arrange an Arrange an Arrange
CHAP'	<u>TER IV</u> <u>Copperethynyl complexes with tertiary</u>	phosphines
	Apparatus and techniques	70
	Monophosphine-ethynyl complexes	71
	Complexes with P:Cu ratio>1	77
	Mixed ethynyl complexes	79
	Phenylbutadiynyl complexes with trialkyl- phosphines	81
	Infra-red spactra	83
	Silverethynyl complexes	84
CHAPT	TER V Gold(I)ethynyl complexes	
	Preparation of phenylethynylgold(I)	85
	Complexes of phenylethynylgold	
	- with amines	86
	- with phosphines	89
	- with triethylarsine and triethylstibine	91
	with isocyanides	91
	Dipole moment data	92
	Reactions carried out to investigate the relative donor strength of various ligands towards phenylethynylgold	95
	Tertiarybutylethynylgold(I) and some derivatives	97
	Reaction between phenylethynylgold and phosphine	99

(iv)

Reaction between quaternary ammonium or phosphonium iodides and phenylethynylgold 100

DISCUSSION

CHAPTER VI Discussion		
Copper(I) Silver(I of monoacetyler) and Gold(I) derivatives les	102
Ethynylcopper(I) of phosphines	complexes with tertiary	105
Ethynylsilver(I)	omplexes	114
Co-ordination comp	oounds of gold(I)ethynyls	114

CHAPTER V	II Interpretation and Discussion of Infra	-red Spectra
The	acetylenic stretching frequencies	121
Isoc	yanide frequencies in the gold complexes	128
<u>N</u> H	frequencies in the gold-amine complexes	129
Meta	l-carbon frequencies	130

APPENDIX I

Stable	tertiary	phosphine	complexes	of	V	
tri	nethylgold	1				134

APPENDIX II

The	far	infra-red	spectra of	some	gold	
(compo	ounds				139

(v)

INDEX OF TABLES

1.	Melting points and solvents of recrystallisation of (RC:C) ₂ Hg	2000 - 2010 - 100 - 100 - 100
2a.	Degree of association of monophosphine (copper ethynyl) complexes	106
2b.	Degree of association, cryoscopically, of (copper ethynyl) bisphosphine complexes	109
2c.	Association, cryoscopically, of PhC:CCuBu ^t C:CCu2PMe ₃	111
2d.	Association, cryoscopically, of phenylbutadiynyl complexes, PhC:C·C:CuL	113
3•	Dipole moments of some phenylethynylgold(I) complexes	115
4.	Stretching frequencies of the acetylenic group in RC:CM	122
5.	Acetylenic stretching frequencies of ethynylcopper(I) complexes	124
6.	Effect of P:Cu ratio on $\mathfrak{J}(C:C)$ in benzene solution of RC:CCu(PMe ₃) _n	125
7.	Acetylenic stretching frequencies in complexes PhC:CAuL	12 7
8.	Acetylenic stretching frequencies in Bu ^t C:CAu and its complexes	128
9.	N-H Stretching frequencies in the complexes PhC:CAu amine	129
10 (a	a) Metal-ethynyl stretching frequencies in RC:CM	131
(1) Metal-ethynyl stretching frequencies in RC:CM-L	132
11.	Infra-red Au-P, Au-Cl and Au-Br stretching	רער

ABSTRACT

Ethynyl copper(I) tertiary phosphine complexes $RC:CCu(PR'_{3})_n$ have been prepared from the familiar polymeric ethynyls $(RC:CCu)_x$. When n = 1 these complexes are dimeric in nitrobenzene and rather more associated in benzene. When n = 2 they are monomeric and one unstable complex with n = 3 has been isolated. A mixed ethynyl complex, $Bu^{\dagger}C:CCu$ PhC:CCu·2FMe₃ and some diacetylene complexes, PhC:C·C:CCuPR'₃ which are similarly associated and the complex (PhC:C·C:CCu)₂(PEt₃)₃, monomeric in nitrobenzene, have also been isolated. One silver complex PhC:CAgPMe₃, dimeric in boiling benzene was prepared.

The first gold(I) ethynyls, other than Au₂C₂ have been prepared. Phenylethynylgold(I) resembles phenylethynylcopper in appearance and in its insolubility; t-butylethynylgold is tetrameric and can be assigned a square structure. They form isolatable complexes RC:CAuL with a wide range of donor ligands, (L), and towards phenylethynylgold the order of donor strength

 $PPh_3 > PEt_3 > P(OR)_3 > RNC > AsEt_3 > SbEt_3 > amines$ has been established.

With the exception of the amine complexes, which appear to be associated in some way, they are monomeric and have dipole moments in the range 4.4 -7.1 D.

Infra-red spectra show that in the gold complexes the

(vii)

(viii)

acetylenic stretching vibration is similar to those of ethynyl groups σ -bonded to various metals but in the tetramer $(Bu^{t}C:CAu)_{4}$ and the polymer $(PhC:CAu)_{x}$ a reduction in frequencies of about 175 cm.⁻¹ indicates π bonding between ethynyl groups and gold atoms which is stronger than in the ethynylsilvers $(RC:CAg)_{x}$ in which the frequency is lowered by about 100 cm.⁻¹ but not as strong in the ethynylcoppers $(RC:CCu)_{x}$ in which it is lowered by 200 cm.⁻¹. In the case of the ethynylcopper(I) compounds, complex formation raises $\lambda(C:C)$ by nearly 100 cm.⁻¹ from that in $(RC:CCu)_{x}$ and in the ethynylsilver(I) compounds by 20 cm.⁻¹. Absorptions due to ethynyl-metal stretching vibrations

were observed in the various compounds for the most part in the range 510-540 cm⁻¹.



INTRODUCTION

The new complexes of Group IB metals described in this thesis are of two main types: ethynyl copper(I) complexes with tertiary phosphines of general formula $(R_3P)_x$ $(CuC!CR')_y$, and ethynyl gold(I) complexes with a variety of ligands RC!CAuL. Interest in these compounds arose from the nature of the acetylides $(R.C!CM)_x$ (M = Cu,Ag) whose structures are still unknown, and from recent work in this department¹ in which a preliminary study had shown that they could be depolymerised by some donor molecules containing Group V elements.

The general picture of the co-ordination chemistry of the coinage metals when univalent is considered in this introduction, complexes with halogens, nitrogen, oxygen and sulphur as donors being mentioned briefly, those with phosphorus (and arsenic) and with carbon-metal bonds are described more fully as they are more relevant to the present work. Finally transition metal acetylene complexes are discussed as during the last ten years compounds with several types of acetylene-metal bonding have been prepared.

CO-ORDINATION CHEMISTRY OF COPPER(I), SILVER(I) AND GOLD(I)

The first ionization potentials (shown in the table below) of the coinage metals are much larger than those of the alkali metals, which they scarcely resemble.



First Ionization Li Na K Rb Cs Cu Au Ag Potential (electron-volts) 5.39 5.14 4.34 4.18 3.90 7.72 7.57 9.22

It is not surprising therefore that they tend to form mostly covalent compounds and the tendency is in the sense Au > Cu > Ag.

The co-ordination numbers exhibited by a metal atom, in a particular valency state, depend on the nature of the ligand and the type of bonding it produces and to a lesser extent on steric In the case of copper(I), silver(I) and gold(I) the effects. preferred co-ordination numbers are usually considered to be: for copper 2 and 4, silver 2 and 3, and gold 2, though compounds have been isolated in which each metal exhibits any one of the states 2, 3 or 4. Examples are described later. Although silver has been thought to have a preferred co-ordination number of 2 and sometimes 3, Arhland and Chatt and co-workers' have found from studies of stability constants of a variety of water-soluble ligands to silver(I) ion that 4 is found for more types of atoms than any other co-ordination number. In particular they clearly demonstrated that silver(I) changes its co-ordination number according to the nature of the ligand. With ligands which afford little or no π -bonding such as ammonia, chloride, bromide and cyanide, considerable stability is shown by the linear 2 coordinated state, though further molecules are taken up in solution to give co-ordination numbers of 3 and 4. With sulphides and selenides, ligands with a slightly greater π -bonding capacity.

complex formation in solution passes smoothly to give 4 co-ordinated silver, and with ligands which form strong π bonds, in particular phosphorus and also to a lesser extent arsenic and iodide ion, co-ordination of 1 and 3 ligands seems to produce the most stable complexes, the stability being increased relative to AgL₂. The tendencies of co-ordinated atoms to form \underline{d}_{π} bonds with silver increases in the order amines < selenides \simeq sulphides < arsines < phosphines and for the halogens in the sense Cl < Br < I.

In the 4 covalent state sp³ tetrahedral hybrid orbitals of the metal are used to accommodate the electrons from the ligand, in the 3 covalent state sp² trigonal hybrid orbitals. The 2 covalent state which on the molecular orbital theory is considered to involve sp hybridisation of the metal, has recently been discussed by Nyholm⁴ and Orgel³. In a recent lecture Nyholm⁴ considers the energy of separation of the <u>ns</u> and <u>np</u> orbitals of copper, silver, and gold, which are 3.79, 3.66 and 4.63 ev. respectively, and shows that the formation of sp^3 tetrahedral hybrid bonds is more difficult for gold(I) than for copper(I) or silver(I). Digonal sp hybridisation is consistent with a large energy difference between the ng and ng orbitals, but in fact the $\underline{s} - \underline{p}$ separations in Cu^+ , Ag^+ , Au^+ and Hg^{2+} are not much greater than in related ions (e.g. Cd^{2+} , Zn²⁺) which do not form <u>sp</u> bonds. For this reason and because the energy difference between the $nd^{9}(n + 1)s$ configuration is

small (for Cu⁺ 2.7, Ag⁺ 4.8, Au⁺ 1.9; cf. Zn^{2+} 9.7 ev.) Orgel³ suggests that hybridisation between the \underline{nd}_{z}^{2} and $(\underline{n} + 1)\underline{s}$ orbitals takes places. This leads to strong electronegativity along the z axis and therefore strong 2-covalent bonding occurs along this axis. Nyholm⁴ has suggested that the residual charge on the metal determines which is the better explanation of 2-co-ordination. If the charge is small, then the larger $\underline{s} - \underline{p}$ separation seems the determining factor and \underline{sp} hybridisation is a reasonable assumption; on the other hand, if the charge is appreciable the smaller $\underline{d} - \underline{s}$ separation and Orgel's picture is the more likely explanation.

Complexes with halogens, nitrogen, oxygen and sulphur

<u>Halogen complexes</u> of the type $[MX_2]^-$ and $[MX_3]^{2-}$ (X = Cl, Br, I), are the most common species in the case of copper and silver⁵. $[MX_4]^{3-}$ complexes are known but are less common. Several crystalline salts containing the $[CuI_2]^-$ anion have been described, for example, $[Cu(en)_2] [CuI_2]_2$. <u>Ammonia</u> forms ammines $[M(NH_3)_n]^+$ with all three elements, where n is usually 2, but can be 1 and 3. Pyridine and substituted ammonias behave similarly.⁶ A variety of nitriles form complexes with copper(I)⁷ and silver(I)⁸ compounds forming mainly 1:1 adducts, e.g., CH₃CNCuCl.

<u>Sulphur</u> has a much greater affinity for these metals than <u>oxygen</u> which forms few complexes. The thioether complexes⁹ of gold(I)

are well known, e.g. monomeric Et_2 SAuCl, and all three elements give a series of thiourea and ethylenethiourea complexes.¹⁰ Polymeric mercaptides of unknown constitution, of copper(I)¹¹ and silver(I)¹² (formula R·S·M), have been prepared and their reaction with aryl halides to prepare thioethers investigated. Most of the chelated complexes of these metals are compounds with oxygen and sulphur as donor atoms. Examples are the complex thiosulphate anion¹³ [Cu(S₂O₃)₂]³⁻ and the copper and gold derivatives of di(methylthio)ethane,¹⁴



Phosphorus and Arsenic Complexes

Copper-phosphine complexes of formula $CuX.nPH_3$ (X = Cl, Br, I; n = 1 or 2) have been isolated. The 2:1 complexes lose phosphine readily to give the 1:1 complexes¹⁵. The only silver and gold complexes with phosphine are those of silver iodide which forms AgI.PH₃ and 2AgI.PH₃, and the very unstable gold compound AuI.PH₃.

Mann, Purdie and Wells⁹, examined the tetrameric tertiary phosphine (and arsine) complexes $[R_{3P}(As) - Cu(Ag)I]_4$ and the

5•

crystal structures of some of them have been investigated.¹⁷ The copper and silver atoms are 4 co-valent being attached to one Group V donor atom and three iodine atoms. Gold(I) halides on the other hand form characteristic monomeric complexes R_3PAuX^9 , many of which are low melting and can be purified by distillation under reduced pressure. Addition of halogens gives the corresponding gold(III) compound, R_3PAuX_3 .

Some 3 covalent copper and silver iodide complexes have been isolated, e.g. $(Me_2NC_6H_4PMe_2)_2CuI$. The analagous gold complexes are ionized.¹⁸

Cotton and Goodgame have recently described some triphenylphosphine complexes in which four phosphine-copper or silver bonds are present.¹⁹ They are isolated as nitrates or perchlorates, e.g. $[Cu(PPh_3)_4]^+ ClO_4^-$. In the case of copper, attempts to isolate the corresponding nitrate yielded instead a bistriphenylphosphine complex $Cu(PPh_3)_2NO_3$, which is not a salt. The nitrate group is apparently involved in coordination to the copper atom, in either a monomer or a nitrate bridged dimer.

Various tertiary arsine complexes have been prepared. Copper(I) halides form complexes with one, two, three or four molecules of methyldiphenylarsine co-ordinated to the copper atom.²⁰ A 4 co-ordinate gold diarsine complex has also been described,²¹ [Au(diarsine)₂]⁺I⁻.

Cochran, Hart and Mann²² have prepared chelate complexes

of all three metals which are stable salts.



x-ray analysis shows a tetrahedral configuration of the ligands.

Tertiary phosphites also form complexes with group IB metals. The compounds $(Pr^{i}O)_{3}PMX$ (X = Cl, Br, I; M = Cu, Ag), are approximately trimeric in solution, and a bisphosphite complex $((PhO)_{3}P)_{2}CuX$ was found to be slightly less than monomeric²³ The analagous gold complexes²⁴ (RO)₃PAuX are monomeric as expected, and can be prepared from phosphorus trichloride-(trichloro)gold AuCl₃.PCl₃, by hydrolysis with an alcohol²⁵ APCl₃.PCl₃ itself is formed by the addition of phosphorus trichloride to an ethereal solution of gold(III) chloride. Verkade and Piper²⁶ have recently used the unusual ligand $P = OCH_{2}^{OCH} C(CH_{3})$

(referred to as 'Phos', by the authors), to prepare some further examples of co-ordinatively saturated complexes. Thus gold gives the dimeric complex $[Au(Phos)_2Cl]_2$ and whereas Cotton¹⁹ could only obtain two phosphine molecules in the coordination sphere in the case of the nitrate complex $Cu(PPh_3)_2$

 NO_3 , 'phos' affords the salt $[Cu(Phos)_4]NO_3$. This ligand has a low steric hindrance and although it is a weaker donor than trimethylphosphite,^{26a} it is probably a better π acceptor.

Cyanide Complexes

The cyanide complexes of copper, silver and gold are well-defined and provide good examples of the characteristic co-ordination of the univalent metals. The salts $M[Cu(CN)_2]$, $M_2[Cu(CN)_3]$ and $M_3[Cu(CN)_4]$ have been isolated, and the main species in solution shown²⁷ to be $Cu(CN)_4^{3-}$. The structure of $KCu(CN)_2$ has been shown to contain spiral polymer chains $[Cu(CN)_2]_{=0}$:



In the case of silver the $Ag(CN)_4^{3-}$ ion is unstable and only $M[Ag(CN)_2]$ and $M_2[Ag(CN)_3]$ have been isolated. The only gold complex is the linear $An(CN)_2^{-}$ ion which is more stable than its copper and silver analogues. $K[Ag(CN)_2]$ has also been shown²⁹ to contain linear ions [NC-M-CN]⁻. With 2:2 bipyridyl $K[Au(CN)_2]$ forms a 4 covalent bipyridyl complex.³⁰

Isocyanide Complexes

Isocyanide complexes have recently been reviewed by Malatesta.³¹ The affinity of the group IB metals towards isocyanides increases with atomic number, though examples of the gold(I) complexes were the last to be prepared. The reaction between copper(I) cyanide and alkyl iodides was investigated by Guillemard³² and later by Hartley.³³ Two series of complexes, with two and three molecules of isocyanide respectively are formed, e.g., CuCN(CNp-tolyl)₃ and CuCN(CNR)₂. Copper(I) chloride and p-tolylisocyanide form the complexes CuCl(CN p-tolyl)n where n can range from 1 - 4. When n = 4the compound is a uniunivalent electrolyte.

The reaction between silver cyanide and methyl iodide takes place in three stages:³⁵

$$2AgCN + CH_{3}I \xrightarrow{\text{room}} 2AgCN \cdot CH_{3}I$$

$$2AgCN \cdot CH_{3}I + CH_{3}I \xrightarrow{40^{\circ}} (AgCN \cdot CH_{3}I)_{2} + \frac{CH_{3}I}{2}$$

$$AgCN \cdot CH_{3}I + AgCN \xrightarrow{100^{\circ}} AgCN(CNCH_{3}) + AgI$$

The alkyl isocyanide complexes are colourless and crystalline but the aryl analogues are rather unstable.³⁶ Klages and coworkers have recently prepared the salts $[A_g(CN p - tolyl)_4]X$, $(X = NO_3, ClO_4^-)$. The nitrate loses isocyanide in ether-chloroform solution to give a bisisocyanide complex $[A_g(CN p - tolyl)_2]NO_3$. Several arylisocyanide derivatives of gold are known.³⁷ The reaction between excess isocyanide and chloroauric acid affords the co-ordinatively saturated gold(I) compound, $[Au(CNR)_4]Cl$, which although stable in the solid state is unstable in solution unless excess isocyanide is present. The corresponding complex with the tetraphenylborate ion $[B(C_6H_5)_4]^-$ is a stable uniunivalent electrolyte. Recrystallisation of tetrakisarylisocyanidegold(I) compounds from ethanol in the absence of excess isocyanide affords the corresponding bisisocyanide complexes (RNC)₂AuX which are probably dimeric with halogen bridges.

Carbon Monoxide Complexes

Copper, silver, and gold do not appear to form stable carbonyls. A reported preparation³⁸ of $Cu(CO)_3$ was later withdrawn as the compound was later found to contain halogen. However small quantities of an unstable copper carbonyl were formed when chloride free carbon monoxide was passed over heated copper at 500°, since a copper mirror was deposited on a colder part of the tube.

Copper(I) halides form carbon monoxide complexes in the presence of water, or ammonia and various amines, e.g. amiline, pyridine.⁴⁰ The compound CuCl.CO.H₂O can be isolated as colourless crystals which lose carbon monoxide in air. It

appears to be dimeric:



The corresponding silver compound has not been isolated, though silver sulphate in fuming sulphuric acid, absorbs carbon monoxide at 0°C. and liberates it on warming.⁴¹ Copper(I) salts prepared by the action of copper(I) oxide On ammonium salts in liquid ammonia,

 $Cu_20 + 2NH_4^+ + 2NH_3 \longrightarrow 2[Cu(NH_3)_2]^+ + H_20$

react with carbon monoxide in liquid ammonia, to yield unstable addition compounds $CuX.CO._{x}NH_{3}$.⁴² Chloro(carbonyl)gold(I) was first prepared⁴³ from carbon monoxide and solid gold(I) chloride at -90°. It is more easily prepared⁴⁴ by bubbling carbon monoxide through a solution of gold(III) chloride in tetrachlorethylene at 100 - 140°.

 $AuCl_3 \longrightarrow AuCl + COCl_2 \longrightarrow ClAuCO$

It crystallises from petroleum as colourless plates and is monomeric in benzene. It is decomposed by water and alcohols. The C:0 stretching frequency 2152 cm.⁻¹ is one of the highest

recorded for metal carbonyls, suggesting that $d\pi - \pi g$ bonding between gold and the carbonyl group is relatively weak.⁴⁵

Alkyls and Aryls

Group IB alkyls and aryls are unstable, the aryls being slightly more stable than the alkyls.

Copper

Methylcopper can be prepared as an unstable yellow solid by the reaction between copper(I) iodide and methyl-lithium at -15° . It decomposes in boiling ether with the formation of copper methane and ethane. In the presence of excess methyllithium (1 mole), it dissolves to an almost colourless solution which probably contains the salt Li⁺[CuMe₂]^{-.46} Methylcopper has also been prepared from copper(II) nitrate and tetramethyllead and can also be detected in the reaction between methylchloride and copper at 350° . At 250° , its half life is 0.002 seconds.⁴⁷

Ethylcopper⁴⁸ is less stable, and a solution in ether decomposes above -50°. The kinetics of the decomposition of methyl and ethyl copper have been studied.⁴⁹

<u>Phenylcopper</u> separates as a greyish brown powder when copper(I) iodide is added to phenyl-lithium. It decomposes at 80° to copper and diphenyl. It is insoluble in most organic solvents but dissolves in pyridine and benzonitrile, and is hydrolysed by

water to benzene and copper(I) oxide. Phenylcopper undergoes l:4 addition to the conjugated system C:C.C:O, in contrast to the l:2 addition of phenyl-lithium. The action of copper powder on diazonium borofluorides at 80° has been reported to yield phenyl, <u>p</u>-tolyl, and <u>o</u>-and <u>p</u>-nitrophenylcopper, though they were isolated only in the form of alleged pyridine complexes, $\operatorname{ArCupy_4}^{50}$.

<u>Triethylphosphinecyclopentadienylcopper</u>, 51 C₅H₅CuPEt₃

This phosphine - stabilized compound is obtained from the reaction between cyclopentadiene, copper(I) oxide and triethylphosphine. It is the most stable organocopper compound, other than some of the ethynyl complexes, decomposing slowly in air. It is stable to water, but hydrolysed by dilute acids to cyclopentadiene. With iron(II) chloride in tetrahydrofuran it gives ferrocene and it seems to be a sigma bonded cyclopentadienyl complex.

Silver

<u>Methylsilver</u> is precipitated⁵² when alcoholic silver nitrate is added to alcoholic tetramethyllead at -10° to -60° .

 $AgNO_3 + Me_4Pb = Me_3PbNO_3 + MeAg$

It decomposes by the formation of ethyl radicals⁴⁹. A more recent study shows the precipitated material to have the composition (MeAg)₂AgNO₃, but with a deficiency of silver

nitrate the product is methylsilver.

<u>Isobutenylsilver</u> is obtained from isobutenyltriethyllead and ethanolic silver nitrate, as an orange solid which affords⁵³ isobutenyl radicals on decomposition above 30° . Similar reactions between Et_3PbR where R is an aryl or alkyl radical afforded RAg though the latter was usually unstable. Its stability is apparently determined by that free radical R· the stability of the silver compound increasing as the stability of R decreases. Thus phenylsilver is relatively stable but benzylsilver is only stable below -100° .

Impure <u>phenylsilver</u> has been isolated as a grey powder by addition of silver chloride or silver bromide to phenyl magnesium bromide at low temperatures. It decomposes at -18° to silver and diphenyl, is explosive when dry; and is insoluble in most organic solvents.^{48,54}

The action of alcoholic silver nitrate on ethyltriphenyllead, ethyltriphenyltin or triphenylbismuth affords a double compound (PhAg)₂AgNO₃ which slowly decomposes at room temperature.⁵⁵

Attempts to isolate phosphine stabilized silver alkyls and aryls by the reaction between a phosphine silver halide complex with a Grignard or organo-lithium reagent have been unsuccessful.⁵⁶ Gold

Gold(I) alkyls or aryls have not been isolated. Evidence for an intermediate gold(I) aryl was obtained by Kharasch and Isbell⁴⁴ who showed that an ethereal solution of chloro(carbonyl)- gold(I) reacts with a Grignard reagent according to the equation:

 $2RMgBr + 2ClAuCO = 2Au + R_2 + 2MgClBr + 2CO$

the more electronegative the radical R, the slower is the formation of gold.

The only organogold(I) complexes prepared before those described in this thesis, were the co-ordination stabilized complexes of the type R_3PAuR' (R = Et, $Ph; \neq Me$, Et, Ph, $PhC:C.,mesityl).^{57}$ They are isolated as crystalline solids (or occasionally as oils) from the reaction between the phosphine gold halide R_3PAuX and the appropriate organo-lithium reagent. They are monomeric in benzene, with dipole moments between 5.5 and 6.5 D.

Gold(III) alkyls and their co-ordination compounds with donors of the oxygen and nitrogen type, have been reviewed.^{58,59}

Ethylene complexes

The stability of the ethylene complexes of group IB metals decreases in the series Cu > Ag > Au, and in fact no ethylene complexes of gold are known.

Copper(I) chloride⁶⁰ absorbs ethylene under pressure and the compound C_2H_4CuCl has been isolated⁶¹. Reaction between butadiene and copper(I) halides affords the pale yellow complex C_4H_6CuX (X = Cl, Br), which can be stored at -78° for several days but decomposes at room temperature or on exposure to air.⁶² Studies on the chemistry of $copper^{63}$ and $silver^{64}$ - olefin complexes in solution have shown that only 1:1 complexes are formed.

Cyclo-octa 1:3 diene and 1:4 diene form unstable addition compounds with silver nitrate, $C_8H_{12}(AgNO_3)_2$. The chelating 1:5 diene gives a more stable 1:1 complex $C_8H_{12}AgNO_3$, m.p., 135-136.⁶⁵

H. J. Lucas and co-workers suggested that the structure of silver-olefin complexes can be represented as a resonance hybrid of the forms:⁶⁶



The bonding has more recently⁶⁷ been regarded as similar to that in palladium and platinum⁶⁸ complexes. The silver 5<u>s</u> orbital interacts with the olefin π u bonding orbital and back co-ordination occurs from the silver <u>d</u> or hybridised <u>dp</u> orbital * to the olefin π g antibonding orbital.



Several complexes between silver perchlorate and olefins are described 69 and are generally of the type (olefin)₂ AgClO₄. Many are explosive, and are decomposed by water.

TRANSITION METAL ETHYNYL COMPLEXES

Reactions of ethynylic molecules with transition metal complexes have provided complexes in which various types of acetylene-metal bonding occur. Many reactions have recently been studied between metal carbonyls and various acetylenes⁷⁰ and a range of rather unusual molecular configurations have resulted. They are not considered here since group IB carbonyl complexes are so few and unstable.

The most common type of ethynyl complexes are those in which there is sigma bonding between the ethynyl group and the metal and also substantial back co-ordination between the occupied metal <u>d</u> orbitals and the acetylene unoccupied antibonding orbitals. Firstly other types of ethynyl complexes will be considered.

'Olefin-type' complexes with disubstituted acetylenes

Ditertiary butyl acetylene reacts with sodium tetrachloroplatinate(II) in ethanol to give the fairly stable compound:



A series of disubstituted acetylene complexes of this type⁷¹ have been prepared but to obtain compounds sufficiently stable to be isolated it seems that there must be at least one tertiary carbon atom adjacent to the triple bond. The acetylene stretching frequency in these complexes is lowered by about 200 cm.¹ (in comparison with 150 cm.¹ in the corresponding olefin complexes) with which the bonding seems to be similar.

The copper complex MeC:CMe,CuCl, which is formed between but-2-yne, and copper(I) chloride has been studied crystographically.⁷² The copper atom lies on the perpendicular bisector of the triple bond. The corresponding silver nitrate complexes with but-2-yne and pent-2-yne have also been isolated⁷³ but are unstable.

Complexes with an acetylene acting as a bidentate ligand

A series of stable ethynyl complexes have been prepared by reducing <u>cis</u>-bistriphenylphosphine(dichloro)platinum(II) with hydrazine in the presence of both mono- and di-substituted acetylenes.⁷⁴

 $\underline{cis}(PPh_3)_2PtCl_2 + N_2H_4 + 2C_2R.R! = 2(Ph_3P)_2PtC_2RR! + N_2 + 4HCl_2PtC_2RR! + N$

The stability of these complexes increases with the electron attracting power of the acetylene, and diparanitrophenylacetylene will replace <u>o</u>-phenylene<u>bis</u>dimethylarsine from its complex $Pt(PPh_3)_2C_6H_4(AsMe_2)_2$. The structure of these complexes can be formulated:



This is concluded from x-ray studies of the dip-bromophenylacetylene complex which suggests a \underline{dsp}^2 square planar configuration of the platinum, and from the I.R. absorption band at 1700 cm⁻¹ which is consistent with an olefinic rather than an acetylenic stretching frequency. The large dipole moments of the compounds in which R_1 and R_2 are both <u>parasubstituted</u> phenylgroups are further evidence that the aryl groups are not co-linear.

Complexes with o-bonding between the metal and the acetylene

Sigma bonded ethynyl groups are present in some compounds of nickel, palladium and platinum of the type⁷⁵ $(R_3P)_2M(C:CR')_2$ (M = Ni, Pd, Pt, R = H, Me, Ph). Their marked stability is attributed to interaction between the metal \underline{d}_{xy} orbital and the acetylene antibonding *m*-orbitals giving the metal carbon bonds appreciable double bond character. Addition of diethylphenylphosphine to the complex with this phosphine gives the 5 co-ordinate complex (PhEt₂P)₃Ni(C:CPh)₂ which is probably monomeric.

Nast and his colleagues have prepared complex anionic ethynyl derivatives of copper^{76,77}, chromium⁷⁸, manganese⁷⁹, iron⁸⁰, cobalt⁸¹ and nickel⁸² by the reaction in liquid ammonia between an alkali metal acetylide and the soluble thiocyanate or nitrate of the transition metal. For example Iron(II) and cobalt(II) form two series of complexes $M_2[Fe(C:CR)_6]$ and $M_3[Fe(C:CR)_6]$ (R = H, Me; M = K, Na), chromium(III) forms an explosive complex $K_3[Cr(C:CH)_6]$. They are generally similar to the iso-electronic cyanide complexes except that they are less thermally and hydrolytically stable. Nickel(II) salts analogous to $Ni(CN)_4^{2-}$ are those of the type $M_2[Ni(C:CR)_4]$ (R = H, Me, Ph), the phenylethynylcomplexes being the most stable. Various other nickel complexes have been prepared including a binuclear complex K₆[Ni₂(C:CR)₈] Electron flow from the d orbitals of the transition metal to the mg antibonding orbitals of the ethynyl groups will oppose the charge transfer to the nickel atom due to the sigma carbon to metal bond, and so decreases the charge on the metal atom and increases the metal carbon order.

Copper(I) and silver(I) ethynyl complexes

Several anionic ethynyl complexes of copper have been isolated by reactions in liquid ammonia, similar to those described above. Copper(I) iodide and propynyl- or phenylethynylsodium afford salts of the type M2[Cu(C:CR)3] and $M[Cu(C:CR)_2]$ (M = Na, K; R = Me, Ph)⁷⁶. These anions are comparable with the $[Cu(CN)_2]^-$ and $[Cu(CN)_3]^{2-}$ ions, but no analogue of $[Cu(CN)_A]^{3-}$ has been reported. The reaction between KC:CH and copper(I) iodide at -78° in liquid ammonia gives the orange coloured copper derivative, CuC:CH. 77 Above -45° it decomposes into a black crystalline copper acetylide. The anionic complexes are not explosive or pyropheric, but are suseptible to hydrolysis by water or acids: $[Cu(C:CR)_3]^{2-} + 2H_2O \longrightarrow CuC:CR + 2RC:CH + 2OH'$ $[Cu(C:CR)_3]^{2-}$ + 3HCl \longrightarrow CuCl₃'' + 3RC;CH Conductivity measurements in nitrobenzene show them to be moderately strong electrolytes. Some will crystallise from ammonia as ammines, e.g. Na[Cu(C:CPh)2]2NH3 which are only stable at room temperature in an atmosphere of ammonia.

Copper(I) and silver(I) carbide and the acetylides RC:CCu(Ag) were familiar compounds in the last century but their structures are still unknown. All the copper derivatives are obtained as bright yellow or red powders, generally insoluble in all solvents with which they do not react, and are apparently polymeric. An interpretation of their structure has been suggested¹ in which the metal atom forms two sigma <u>sp</u> hybrid bonds, an acetylene group acting as a donor to provide an extra electron.

Back co-ordination is expected from the fitted metal <u>d</u> orbitals to the Ig antibonding orbitals of at least two acetylene groups. Additional bonding between layers should be possible if the polymer chains are arranged so that acetylene groups lie above and below the copper atoms. The corresponding colourless silver derivatives presumably could have a similar structure. In many cases where R is an alkyl group the silver compounds are moderately soluble in several organic solvents, in particular in chloroform, carbon tetrachloride and acetone.⁸³ R. Vestin and E. Ralf⁸⁴ have reviewed the earlier silver acetylide chemistry.

A few silver complexes with disubstituted acetylenes are known 69,85 . Silver nitrate forms a crystalline complex with

diethylacetylene, $C_6H_{10}AgNO_3$ and silver perchlorate is soluble in diethyl and methylethylacetylene, the solid complexes $C_5H_8(AgClO_4)_3$ and $C_6H_{10}(AgClO_4)_3$ having been isolated. A 5% solution of silver nitrate in 95% ethanol gives a quantitative precipitate of even traces of 1-alkynes as the silver acetylide-silver nitrate complex RC:CAg, AgNO₃. Treatment with aqueous sodium cyanide regenerates the acetylene. The insolubility of the copper acetylides has long been used to detect and purify monoacetylenes. Both the copper and silver compounds readily yield the acetylene with dilute mineral acids or sodium cyanide solution⁸?

Copper(I) ethynyl complexes, RC:CCu, are oxidised to diacetylenes RC:C-C:CR in aqueous copper(I) chloride ammonium chloride solution. This coupling reaction is also catalysed by copper(II) chloride, hydrogen peroxide and potassium ferricyanide.⁸⁸ The oxidation is best carried out in the presence of a base⁸⁹. The copper catalysed reaction of mono-substituted acetylenes in aqueous pyridine has been studied kinectically which suggests the formation of intermediate ethynyl anions.⁹⁰

Phenylethynylcopper(I) forms a colourless unstable ammine PhC:CCuNH₃ which loses ammonia readily in air or in a stream of nitrogen⁷⁶. It is sparingly soluble in liquid ammonia but dissolves as a soluble adduct with carbon monoxide which is too unstable to isolate.⁴² An isocyanide complex PhC:CCu-

 $(p-tolyl NC)_3$ has been isolated ³⁴ as white crystals which decompose above 100°. The first co-ordination compound with phosphorus as a donor was prepared by G. Calvin⁹¹ and G. E. Coates of this department who obtained PhC:CCuPEt, as yellow needles, soluble in hexane and many organic solvents. Subsequent work^{1,92} showed the donor strength of group V elements to R.C:CCu to be in the order $N \ll P \gg$ As and that arylethynyl complexes are stabler than alkynyl complexes. Thus propynylcopper was found to be unaffected by a variety of amines and to form a weak complex with triethylphosphine, which decomposed immediately in air, whereas phenylethynylcopper dissolves in pipendine and isopropylamine and the triethylphosphine complex decomposed only slowly. The latter complex was found to be dimeric in nitrobenzene and slightly more associated in benzene. The analogous silver complex PhC:CAgPEt, was found to be similarly associated in solution and the structure



was suggested.

Phenylethynylsilver(I) in contrast to methylethynylsilver dissolves in various amines and an unstable isopropylamine complex which loses amine on exposure to air, and a slightly more stable triethylarsine complex were isolated. It appears that the silver complexes are slightly more stable than their copper analogues.

A more comprehensive further study of ethynylcopper complexes has been carried out (and is now described) with a view to (a) obtaining more extensive molecular weight data on various ethynylcomplexes, (b) finding out whether more than one molecule of tertiary phosphine would co-ordinate to each copper atom, (c) investigating the infra-red absorption frequencies of the ethynyl groups in these compounds.

Gold(I) ethynyl complexes

Gold(I) acetylide, Au_2C_2 , was known sixty years ago, having been obtained as an explosive powder.⁹³ It is surprising that no other gold acetylides have previously been described.

The only gold ethynyl complex reported⁵⁷ is PhC:CAuPEt₃ prepared in a series of experiments mentioned earlier. It is monomeric, in contrast to the analogous silver and copper compounds, gold(I) showing its characteristic tendency not to raise its co-ordination number above two.
EXPERIMENTAL

CHAPTER I

PREPARATION AND PURIFICATION OF REAGENTS

The attempted preparation of trialkylphosphines from trialkylaluminiums

The reaction

 $AlR_3 + PCl_3 \longrightarrow RPCl_2 + R_2PCl + R_3P$

has been reported⁹⁴ to give alkyldichlorphosphines in satisfactory yields, and using excess triethylaluminium a 30% yield of triethylphosphine has been obtained.

Attempts were made to prepare trimethyl and triethylphosphine using aphosphite instead of phosphorus trichloride. A reaction of this type, if successful, would obviate the use of large amounts of ether necessary in Grignard or alkyl lithium methods of preparing tertiary phosphines. Unfortunately the most successful reactions afforded no more than a 9% yield of trimethyl or triethylphosphine. One of these is described.

Since this work was carried out, M. Sander⁹⁵ has shown that on heating the adduct formed between an alkyl phosphite and a trialkylaluminium, the phosphite rearranges to a phosphonate (in 70% yield), in a similar way to the Arbuzov reaction. Preparation of trimethylaluminium^{96,97}

6MeI + 4Al	$= Me_4Al_2I_2 + Me_2Al_2I_4$
$3Me_4Al_2I_2 + 6Na$	= $6NaI + 2AI + Me_6Al_2$

Aluminium turnings (80g., 3 mole) were added to dimethylaluminium iodide (60g.) from a previous reaction, followed by methyl iodide (568g., 249 cc., 4 mole) over 3 hours, the reaction being kept at 80° with an oil-bath. The mixture was then fractionated, using an electrically heated and well-lagged glass helices column, 50 cm. long, and a take-off head. The fraction boiling $62-63^{\circ}/75-76$ mm., pure trimethylaluminium (20 cc.) which froze in the receiver, and a second fraction $110^{\circ}/50$ mm., mostly dimethylaluminium iodide (297g.) were collected.

The dimethylaluminium iodide was transferred to a dropping funnel and cautiously added, with stirring, to excess sodium (56g.) keeping the reaction at 130° with an oil-bath. An unbreakable mass was obtained, which on distillation gave trimethylaluminium (58g., 0.805 mole, 40%), b.p. 126°. Attempted preparation of trimethylphosphine

Trimethylaluminium (40g., 53 cc., 0+55 mole) in methylcyclohexane (100 cc.) and triphenylphosphite (redistilled, b.p. 178-180[°]/<0.5 mm) (171g., 140.5 cc., 0.55 mole) in methylcyclohexane (260 cc.) were added dropwise with stirring



into a l litre3necked flask containing methylcyclohexane (30 cc.). The flask was connected to a trap cooled to -78°. A slightly exothermic reaction warmed the flask to 60°, and the solution turned cloudy. Attempts to distil out any phosphine were unsuccessful. After cautious hydrolysis with caustic sodasolution, the mixture was steam distilled, and the organic layer treated with silver iodide in aqueous conc. potassium iodide. The silver iodide complex obtained (15g.) corresponded to a 9% yield of trimethylphosphine.

Preparation of trimethylphosphine from methyl-lithium

2Li + CH ₃ Cl	Konese Konasta	LiCl + MeLi
3MeLi + PCl ₃		Me ₃ P + 3LiCl
4 ^{Me} 3 ^P + 4AgI		[Me ₃ P AgI] ₄

(a) Preparation of the silver iodide complex

Lithium shot (70g., 10 mole) was prepared in the apparatus shown (Fig.1.). The petroleum oil (b.p. 210/230) was heated to boiling and the lithium then stirred vigorously with the 'vibromixer' stirrer. The shot was washed through the widebore tap with ether (1200 cc.) into a 5 litre 3 necked flask. Methylchloride was passed through the solution cooled to -5° , the required amount being measured with a previously calibrated flowmeter. The methyl-lithium (4 mole) solution was transferred to two l litre dropping funnels and added to phosphorus(III) chloride (173.5g., 110 cc., 1.26 mole) in ether (500 cc.), at -78° C, allowing 5% excess methyl-lithium. After hydrolysis with deoxygenated water, the organic layer was separated and added with good stirring to silver iodide in concentrated potassium iodide solution. The precipitated silver iodide complex⁹⁸ was filtered, washed with conc. potassium iodide solution and methanol and dried in vacuo. Yield 224g., 0.72 mole, 57%, m.p. dec.> 100° .

(b) Decomposition of the silver iodide complex

Tetrakistrimethylphosphineiodosilver(I) (224g.) was placed in a long-necked flask surrounded by an air-bath. A sintered disc was placed between the flask and a cold trap which in turn was connected to a vacuum line with ampoules for collecting and storing the phosphine. After obtaining a good vacuum, the air bath was heated to 200° , and the phosphine collected in the ampoules cooled to -78° . Yield of trimethylphosphine 45g., 82%. The unchanged complex was recovered and added to that for further decompositions.

In more recent preparations by Mr. R. Coult of this department, the trimethylphosphine has been isolated and stored in 4N sulphuric acid as its air-insensitive salt Me_3HP^+ HSO₄⁻. The phosphine was released as required, by addition of caustic soda solution, into an organic solvent which was convenient for the reaction being studied. The phosphine solution is

dried over magnesium sulphate, and standardised by weighing the meth-iodide formed from a known aliquot.

Preparation of triethylphosphite⁹⁹

3EtOH + PCl₃ + PhNEt₂ = 3PhNEt₂HCl + (EtO)₃P To pure dry ethyl alcohol (143.5g., 182 cc., 3.1 mole) and diethyla^Ailine (445g., 482 cc., 3.05 mole; redistilled 215-216[°]) in dry pentane (2 litres), phosphorus(III) chloride (137.4g., 88.6 cc., 1 mole) was added, over 1 hour, with good stirring at 10-15[°]. The white precipitate was filtered in air, and the filtrate gave, on distillation, triethylphosphite (129g., 82.4%, b.p. 155-160[°]).

Preparation of triethylstibine

Antimony(III) chloride (79.5g., 0.348 mole; redistilled b.p. $158-159^{\circ}$) in ether (250 cc.) was added to ethylmagnesiumbromide (from magnesium 38g.) in ether (500 cc.) at -20° . After most of the antimony(III) chloride had been added, a black colour (finely divided antimony) developed. At this point it would probably have been advantageous to stop, instead of adding the remainder. Hydrolysis with oxygen-free ammonium chloride solution, and separation of the ether layer which was dried and finally distilled gave triethylstibine (40g., 0.192 mole, 58%, b.p. $161-162^{\circ}$).

Preparation of prop-1-yne(methylacetylene) and its copper and silver derivatives.

 $2Na + 2NH_{3} = 2NaNH_{2} + H_{2}$ $CH_{3}CHBr \cdot CH_{2}Br + 3NaNH_{2} = CH_{3}C:CNa + 2NaBr + 3NH_{3}$ $CH_{3}C:CNa + H_{2}O = CH_{3}C:CH + NaOH$

Methylacetylene released by hydrolysis of propynyl sodium was passed directly into ammoniacal copper(I)chloride or silver nitrate to isolate the required copper(I) and silver(I) derivatives.¹⁰⁰

A 2 litre 3 necked flask was filled with liquid ammonia (1200 cc.) and cooled to -40°. Sodium was catalytically decomposed to sodamide¹⁰¹ by adding powdered hydrated iron(III) nitrate (c. lg.) to the liquid ammonia followed by sodium (2g.). The deep blue colour was replaced by the black colour of finely divided iron. Sodium (76g., 3.3 mole) was added in lOg. quantities allowing time for it to react before adding further portions. When no further blue colour remained, l,2-dibromopropane (202g., l mole) was added cautiously to the fine grey suspension of sodamide, a vigorous reaction ensueing. The mixture was stirred for a further hour and the ammonia allowed to evaporate. Two preparations were carried out to this stage.

In the first preparation, the methylacetylene which was now liberated by addition of brine (1000 cc.) was bubbled into ammoniacal copper(I)chloride solution (copper(I)chloride (80g., 0.8 mole) was suspended in water (500 cc.) and ammonia added

till the solid dissolved). This solution was stirred well, a carbon dioxide reflux condenser being attached to the exit neck of the flask to condense any unreacted methylacetylene. When addition of brine was complete the contents of the reaction flask were heated to drive over any methylacetylene that remained. The yellow precipitated propynylcopper was filtered off, washed with ammonia (4N), ammoniacal ethanol, acetone and ether and dried in vacuo. Yield 50g., 0.485 mole, 48.5%. The relatively low yield was due to loss of some material during the vigorous reaction which took place on addition of the 1,2-dibromopropane.

In the second preparation, the methylacetylene was bubbled through ammoniacal silver nitrate (from silver nitrate 153g., 0.9 mole) which afforded after filtration, washing with water, ethanol, acetone and ether and drying, propynylsilver 112.5g., 76%.

<u>Preparation of 3:3 dimethyl-l-butyne (t-butylacetylene)</u> $(CH_3)_3C \cdot CC_{2}CH_3 + 3NaNH_2 = (CH_3)_3C \cdot C:CNa + 2NaCl + NH_3$ $(CH_3)_3C \cdot C:CNa + H_2O = (CH_3)_3C \cdot C:CH + NaOH$

2,2-dichloro3,3-dimethylbutane was dissolved in toluene and dried by adding anhydrous magnesium sulphate. The solution was filtered. Sodium (76g., 3.3 mole) was converted to sodamide using ferric nitrate as described above.

2,2-dichloro3,3-dimethylbutane (143g., 1 mole) in toluene (250 cc.) was added over 11 hours with stirring, and the ammonia was then allowed to evaporate overnight. The residue was hydrolysed with brine (500 cc.) followed by water (500 cc.) the volatile products being passed through. (a) an empty 2 litre flask as safety trap, (b) a 2 litre flask containing water and phenolphthalein, to which hydrochloric acid was added from a dropping funnel to neutralise the ammonia, (c) a calcium chloride drying tower, (d) two traps cooled to -78° to condense the t-butylacetylene. The liquid which had collected in the empty flask and on the surface of the water was separated, dried over magnesium sulphate, and combined with that from the traps and the trap washings with dry toluene. The whole was distilled up a short helices column. The fraction b.p. 37.2-38°, t-buty1acetylene, was collected. Yield 40.7g., 0.495 mole, 49.5%.

Preparation of p-bromophenylacetylene

1.	$C_{6}H_{5}Br + (CH_{3}CO)_{2}O$	AlCl_3>	Brc ₆ H ₄ COCH ₃
2.	Brc ₆ H ₄ COCH ₃ + PCl ₅	\longrightarrow	$\operatorname{Brc}_{6}\operatorname{H}_{4}\operatorname{CCl}_{2}\operatorname{CH}_{3}$ + POCl_{3}
3.	$\text{Brc}_{6}\text{H}_{4}\text{CCl}_{2}\text{CH}_{3}$ + 2KOH	\rightarrow	Brc ₆ H ₄ C:CH + 2KCl + 2H ₂ O

1. Preparation of p-bromacetophenone¹⁰²

Bromobenzene (157g., 105 cc., 1 mole) was added to

aluminium(III)chloride (294g., 2.2 mole) in carbon disulphide (500 cc.). Acetic anhydride (102.1g., 94.4 cc., 1 mole) was added with good stirring, and the contents of the flask refluxed for 30 minutes. The mixture was poured on to icewater, extracted with ether, and the ether layer dried over magnesium sulphate and distilled. Two fractions were collected: (a) 36.7g., 60-114°/7 mm., (b) 136.7g., 114-116°/ 6.75 mm., p-bromacetophenone, m.p. 48-50°, 69%.

2. Preparation of p-bromophenyl2, 2-dichlorethane¹⁰³

Parabromacetophenone (130g., 0.65 mole) and phosphorus(V) chloride (146g., 0.70 mole) were mixed in a 500 cc. 3 necked flask. On warming, when the mixture melted a vigorous reaction took place which was complete in 15 minutes. Distillation gave two fractions: (a) $45-55^{\circ}/18$ mm., mostly phosphorus oxychloride, (b) $128-134^{\circ}/18$ mm., <u>p</u>-bromphenyl-2:2dichlorethane, 117.6g., 0.47 mole, 72%.

3. <u>Preparation of p-bromphenylacetylene</u>¹⁰³

Parabromphenyl2,2-dichlorethane (115g., 0.46 mole) was added to ethanolic potassium hydroxide (129g., in 610 cc. ethanol, 25% solution) and refluxed for three hours. The resulting liquid was poured on to ice, and the oil produced was separated. The aqueous ethanolic layer was extracted with ether, and the extracts combined with the oil and dried over potassium hydroxide. Removal of the ether and distillation

afforded <u>p</u>-bromophenylacetylene (22g., 0.12 mole, 26%), which was recrystallised from methylated spirits, m.p. 59-61⁰ (literature gives⁹ m.p. 64-65⁰).



1. Preparation of 2,4,6-trimethylacetophenone¹⁰²

Mesitylene (51.1g., 47.5 cc., 0.50 mole) was added to aluminium(III)chloride (147g., 1.1 mole) in carbon disulphide (200 cc.). Acetic anhydride (51.05g., 47.2 cc., 0.50 mole) was added with good stirring. The mixture was refluxed for 30 minutes, poured on to ice, extracted with ether and the ether layer dried. Removal of the ether and distillation afforded 2,4,6-trimethylacetophenone, 66.4g., b.p. 86-87°/2.75 mm., 82%.

2. Preparation of 2,4,6-trimethylachlorostyrene¹⁰⁴

2:4:6 trimethylacetophenone (60g., 0.37 mole) was added to phosphorus(V)chloride (90g., 0.43 mole) in a 500 cc. 3 necked flask. The mixture was heated on a boiling water bath for 4 hours, and then poured on to ice. The dark green oily layer was separated, neutralised with 10% caustic soda, and combined with the ether extracts of the neutralised aqueous layer. The ethereal solution was dried over magnesium sulphate and distilled. Two fractions were collected: (a) 124-130[°]/ 25 mm. 2,4,6-trimethyl<u>a</u>chlorostyrene, 43.lg., 0.239 mole, 65%; (b) 132-136[°]/12 mm., 5.4g., - probably the dichlorocompound.

3. Preparation of mesitylacetylene¹⁰⁴

2,4,6-trimethylachlorostyrene (42g., 0.233 mole) was added to a suspension of sodamide (llg., 0.275 mole) in petroleum oil (fraction 210-240°) and the mixture heated, with an oilbath at 150° for 3 hours. After cooling, it was poured on to ice, acidified with dilute hydrochloric acid and extracted with ether. The ether layer was dried over magnesium sulphate. After removal of the oil, distillation afforded two fractions: 85-110°/23 mm. (c. 60 cc.), petroleum oil and mesityl-(a)acetylene which could not easily be fractionated, (b) 110-114°/ 23 mm., (c. 50 cc.), almost entirely petroleum oil. The mesitylacetylene was therefore used as a solution in petroleum oil to prepare the required copper(I), silver(I) and mercury(II) derivatives in the usual way (see page 63). By preparing mesitylethynylcopper from a known aliquot of the petroleum solution the yield of mesitylacetylene was calculated to be

14.4g., 0.076 mole, 33%.

Further reactions should be carried out in a higher boiling oil, e.g. heavy white oil.

Preparation of 1.	-phenylbutadiyne			
DrCH • CHCHO	1. MeMgBr	Doch • CH • CH • CH		
L 11011 • 0110110	2. KHS04	111011-011-011-0112		
PhCH:CH.CH:CH2	$3 \cdot Br_2$	PhC:C·C:CH		
2	4. NaNH ₂	2		

1. + 2. Preparation of phenylbutadiene¹⁰⁵

Cinnamaldehyde (redistilled b.p. 84-86°/1 mm., 80g., 0.67 mole) in ether (500 cc.) was added to methylmagnesiumbromide (c. 0.73 mole, from magnesium (18.5g.)). After hydrolysis with excess ammonium chloride, a further 200 cc. of ether was added and the layers separated. Ether was removed and the residue distilled in the presence of potassium hydrogen sulphate (8g.). Fractionation gave pure phenylbutadiene (55.5g., 70%, b.p. 44-47°/0.5 mm.), which solidified on cooling to 0°. 3. Preparation of phenylbutadienetetrabromide¹⁰⁶

Bromine (106g., 34 cc., 0.66 mole) in carbon tetrachloride (200 cc.) was added to 1-phenylbutadiene (41.5g., 0.319 mole) in carbon tetrachloride (100 cc.). The bromine was decolourised rapidly at first, but then more slowly with formation of the only moderately soluble white tetrabromide. The reaction was stirred for 3 hours before filtering. The filtrate was shaken with 2N sodium carbonate to remove excess bromine, the carbon tetrachloride evaporated, and the solids combined and recrystallised from petroleum ether (60-80°). Yield of recrystallised tetrabromide, 116g., 81%, m.p. 123-124°. A similar preparation afforded a further 145g., 85%.

4. Preparation of 1-phenylbutadiyne¹⁰⁶

Discussion. As reaction of phenylbutadienetetrabromide with sodamide has been reported¹³ to give yields of the unstable diyne from only 8-20%, attempts were made to isolate the required copper derivative by adding the ether extract, obtained after evaporation of the ammonia, to ammoniacal copper(I) chloride and so avoid isolating the diyne itself. However, instead of yellow-orange acetylide, the solid which was obtained was brown and contaminated with the polymeric solid always present in large amounts as a biproduct in this reaction. The preparation is best carried out by using a small amount of the tetrabromide (say 30g.) in any one reaction, and working up the product as quickly as possible. The most successful preparation is described below. Several yields of only 20% were obtained. <u>Preparation</u> The tetrabromide (30g., and 25.5g.) in two concurrent experiments was added, in small portions and with good

stirring≸, to sodamide (from sodium 8g. and 7g.) in two 1 litre neck flasks. Ether was then added which encouraged rapid evolution of the ammonia, followed by brine (2 x 100 cc.). The mixture was filtered to remove the brown insoluble material which was washed well with ether. The ether layers from each experiment were combined and the ether removed, at the pump, to leave a brown oily liquid which on distillation, collecting the distillate in a cold trap, afforded phenylbutadiyne (6.2g., 0.049 mole 40%), b.p. 45°/0.05 mm. which agrees with that given by Schlubach and Franzen¹⁰⁷. Polymerisation to a dark brown mass took place rapidly above 45°. Phenylbutadiyne may be kept at -78°, but above 0° it melts and darkens rapidly. It was converted immediately to the required copper(I) silver(I) and mercury(II) derivatives. Preparation of copper(I) chloride¹⁰⁸

A filtered solution of sodium sulphide $(N_{a_2}SO_3 \cdot 7 H_2O_3)$, 267g., 1.1 mole) in water (750 cc.) was added slowly to a filtered solution of copper(II) chloride $(CuCl_2 \cdot 2H_2O_3, 170.5g.)$, 1 mole) in water (150 cc.). The mixture was poured into a large volume of water(5 litres) containing sodium sulphite (26.7g., 0.1 mole) and conc. hydrochloric acid (20 cc.). The pale blue solution was decanted and the copper(I) chloride filtered, washed with glacial acetic acid, ethanol and ether and dried at 80° for 20 minutes. Yield 94g., 93%. It was stored under nitrogen, in the dark.

The author would like to thank the following for samples of the compounds listed:

Dr.	D.	Blake	ethylacetylene	
Mr .	J.	M. Braddock	dimethylphenylphosphine	
Mr.	R.	Coult	triethylarsine, 2:2 dichloro	3:3-
			-dimethylbutane	
			p-nitrophenylacetylene	
			Triethylphosphine	
			tri-n-propylphosphine	
Mr.	Μ.	Jewitt	diethylphenylphosphine	
Mr.	J.	Lloyd	n-butyl isocyanide	
			o-ethylphenylisocyanide	
			<u>p</u> -tolylisocyanide	

Purification of solvents

Benzene

(a) For Molecular Weight and Infra Red Measurements.

'Analar' benzene was dried over sodium wire (b) For Dipole Moment Measurements

'Analar' benzene (10 litres) was fractionally crystallised, shaking the freezing solution at half-hourly intervals, and discarding about one sixth. This was repeated three times. It was then distilled from potassium and stored in a flask attached to a column containing molecular sieve (Linde 4A), so that about 400 cc. was in contact with the sieve immediately before use.

Ether

"Ether Anhydrous Methylated" (J. Macfarlan & Co., Htd.) was dried over sodium wire for 24 hours.

Methanol and ethanol¹⁰⁹

Magnesium (usually 1 mole) was placed in a 2 litre flask under nitrogen and the alcohol (200 cc.) added. Stirring with gentle warming at first initiated the reaction, and when it had subsided the remainder of the alcohol (1 litre) was added. After refluxing for two hours, the dried alcohol was distilled leaving behind the residue of magnesium hydroxide and alkoxide.

Nitrobenzene

For molecular weight determinations, 'Aźnalar' nitrobenzene (2½ litres) was fractionally crystallised three times (as for benzene). It was then stood overnight over phosphorus pentoxide and distilled from phosphorus pentoxide under reduced pressure.

Tetrachlorethylene 110

Tetrachlorethylene $(2\frac{1}{2})$ litres) was stood over calcium chloride for several days and then fractionally distilled, collecting the fraction with b.p. 119.5°. Thymol (5 p.p.m.) was added as a stabilizer¹¹¹ to avoid oxidation to phosgene and trichloracetic acid, and it was stored under nitrogen in the dark.

Petroleum Fraction, 200-230.

Petroleum (200-230°; 2 litres) was shaken with concentrated sulphuric acid (100 ml. portions) until the acid was only slightly discoloured red. It was then washed with water, sodium carbonate and more water until the washings were neutral. After standing over magnesium sulphate the fraction distilling 210-230° was collected.

CHAPTER II

EXPERIMENTAL METHODS

A. MOLECULAR WEIGHT DETERMINATIONS

Cryoscopically

Molecular weight determinations, in benzene and nitrobenzene (the solvents being purified as described on page 40) were carried out in an apparatus of standard design¹¹², which consisted of the solution tube, with nitrogen inlet and Beckmann Thermometer, surrounded by a second tube as an air jacket, the whole being immersed in a cooling bath at four to five degrees below the freezing point of the solvent. Α known volume of solvent was placed in the inner tube containing 4 pellets of molecular sieve (Linde Grade 5A) and varying amounts of solid were added and the depressions measured. Alternatively solvent (10.00 cc.) was placed in the tube initially. and 5 cc. aliquots of a solution of known strength of the compound added, the latter method being favoured when the compound was very soluble, and large amounts were available. A slow nitrogen stream was passed over the solution to avoid decomposition of air-sensitive compounds and to prevent condensation of water-vapour. The nitrogen flow rate appeared to be critical: if it was too fast evaporation of the solvent occurred, and if too slow condensation of water vapour. Diphenyl was used to



determine the cryoscopic constant for benzene (found to vary between 51.2 and 53.5 for various purifications of benzene) and <u>paranitrotoluene</u> in the case of nitrobenzene (k = 69.6 - 70.9).

Ebullioscopically

A Swietoslawsky ebulliometer, shown in Fig. 2, was placed in a large cardboard box to exclude draughts. Solvent (25.00 cc.) was boiled steadily with a microburner so that the pure solvent flowed/over the well containing thermistor A, and pure vapour refluxed over well B, and then from a water condenser attached A few drops of heavy white oil were used to ensure good at C. contact between the thermistors and the wells. The matched thermistors formed two arms of a Wheatstone Bridge circuit, a variable resistance R, being in series with thermistor A. The ratio arms were each of 1000 ohms, and a/10 volt D.C. transistor supply was used to drive the circuit together with an 8000 ohm resistance so that the heat dispersion of the thermistors was well below the maximum (10 m_watts). A relatively large voltage was chosen, as it could be reproduced satisfactorarily, because it is important that the characteristics of the circuit are the same for both the compound whose molecular weight is required and the standard, as the change in resistance per millimole of solute varies slightly with the small current through the thermistors. Equilibrium was attained after boiling for

45 minutes, and a weighed amount of the compound was then added down the top of the condenser. A second condenser was placed above this, and the water withdrawn from the latter, so that the solvent climbed up the condenser and washed the compound into the apparatus. Water was again passed through the lower condenser and the solution left for 10 minutes to attain equilibrium. The change required in the variable resistance to rebalance the circuit was measured. This increase in resistance is proportional to the elevation of the boiling point which is proportional to the number of moles of solute. Diphenyl was used as a standard and the standard was repeated to ensure reproducibility. A graph (see overleaf) relating the change in resistance per millimole of diphenyl was plotted. Hence by finding the resistance change for weighed amounts of a compound, the number of millimoles present, and therefore its molecular weight can be calculated.

45.

Benzene and acetone were found to be the most suitable solvents though frothing of the solution over both wells occurred at greater concentrations (usually c. 4%). Steady conditions were not achieved with diethylether. Results are collected in Table 2_{α} , page 106.

B. DIPOLE MOMENT MEASUREMENTS

Theoretical

Faraday113 showed that the molecules of a dielectric



when placed in an electric field become polarised. Clausius¹¹⁴ and Moscotti¹¹⁵ interpretated this mathematically and derived the relation

$$\frac{\varepsilon - 1}{\varepsilon + 2} \cdot \frac{1}{d} = p$$

where \mathcal{E} is the dielectric constant, d the density, and p the specific polarisation. Multiplying this equation by the molecular weight M, gives the molecular polarisation P

$$\frac{\varepsilon - 1}{\varepsilon + 2} \cdot \frac{M}{d} = P \qquad (1)$$

The polarisation is constant for many materials of small dielectric, independent of temperature and pressure and varies little between the solid and liquid states. For these materials, the relation between the dielectric constant and the refractive index n, is that developed by Maxwell¹¹⁶, namely $n^2 = \mathcal{E} \emptyset$, where \emptyset is the magnetic permeability of the material and is almost unity for ferromagnetic substances. If the values of the refractive index obtained in the visible region are extrapolated to infinite wavelength, which is virtually the wavelength at which \mathcal{E} is measured, then

In practice in the case of materials of high dielectric

eonstant

 $E > n_{\infty}^2$. From (1) and (2):

$$\frac{n^2-1}{n^2+2} \cdot \frac{1}{d} = r$$

Lorenz¹¹⁷ and Lorentz¹¹⁸ showed this relation to hold, where r is the specific refraction and multiplied by M gives the molecular refraction R.

$$R = \frac{n^2 - 1}{n^2 + 2} \cdot \frac{M}{d}$$
(3)

Molecular refractions are additive within certain limits and can be calculated by allowing appropriate values to the various atoms and bonds comprising the molecule. On the other hand the molecular polarisation P, is only additive for compounds of low dielectric constant (where $\mathcal{E} = n_{\infty}^2$).

This difference in behaviour was explained by Debye¹¹⁹ who suggested that although a molecule as a whole may be neutral there was no reason why the distribution of charges should not be unsymmetrical in some molecules in the absence of an externally applied field, giving a permanent charge separation or dipole which would orientate itself in an electric field of relatively low frequency (i.e. radiofrequency).

The Total Polarisation = Orientation Polarisation (Po) + Distortion or Induced Polarisation.

 $T \cdot P \cdot = P_0 + P_E + P_A$

(4)

where P_E and P_A are the electron and atom polarisations respectively. The atom polarisation arises since if a molecule contains polar bonds, the atoms carry different effective charges and so the nuclei are displaced with respect to one another thus contributing to the induced dipole.

The refractive index is ordinarily measured for visible light, the refraction being due primarily to the displacement of electrons and so equation (3) gives only the electron polarisation.

$$P_E = \frac{n^2 - 1}{n^2 + 2} \cdot \frac{M}{d}$$

The refractive index may be extrapolated to infinite wavelength, but values of R obtained in this way are only slightly different from those at visible frequencies and often no correction is made in calculating dipole moments.

Measurement of the dielectric constant gives the total polarisation, since in the slowly alternating field there is time for molecules to 'line up' and atoms to undergo displacement. P_A cannot be measured except in the case of simple molecules by studying their infrared spectra in which atom polarisations are additive functions of various absorptions,¹²⁰ or in the case of symmetrical molecules for which P_0 is zero. P_A is small and can often be ignored in the case of highly polar compounds, or assumed to be between

5-15% of P_R.

From the Debye Theory in which a molecule with permanent dipole, μ is inclined to an electric field, F the expression

$$P_0 = \frac{4}{3} \pi \frac{Nu^2}{3kT}$$
 is obtained.

(k = Boltzman's constant, T = absolute temperature, N = avogadro number)

$$\therefore \mu = \sqrt{\frac{9kT}{4\pi N}} \cdot P_0$$

As all measurements were made at 25°C,

$$\mu = 0.2212 \sqrt{P_0}$$

And since from equation (4):

$$P_0 = (T_{\bullet}P_{\bullet}) - (P_E + P_A)$$

the dipole moment can be evaluated.

Dipole moments of most molecules lie between 0 and 9×10^{-18} e.s.u. (0-9 Debye), since the magnitude of the electronic charge is 4.0 x 10^{-10} e.s.u. and the distance between small atoms is 1 to 2 x 10^{-28} cm.

The polarisation of a solution may be expressed in terms

of the polarisation $(p_1 \text{ and } p_2)$ and the mole fractions $(f_1 \text{ and } f_2)$ of the solvent and solute respectively as given by Debye¹¹⁹:

$$P = p_1 f_1 + p_2 f_2 = \frac{\varepsilon - 1}{\varepsilon + 2} \frac{M_1 f_1 + M_2 f_2}{d}$$

Halverstadt and Kumler¹²¹ used specific volumes instead of densities and weight fractions instead of mole fractions. By assuming that the specific volume and dielectric constant of the solution are linear functions of the weight fraction (i.e. $\mathbf{E} = \mathbf{a} + \alpha \mathbf{w}_2$ and $\mathbf{V} = \mathbf{b} + \beta \mathbf{w}_2$) where a and b are the dielectric constant and specific colume of the solvent respectively, and \mathbf{w}_2 is the weight fraction of the solute, they derived an expression for the specific total polarisation of the solute at infinite dilution:

$$T \cdot P \cdot 2 = \frac{3\alpha V_1}{(\varepsilon_1 + 2)^2} + \frac{\varepsilon_1 - 1}{\varepsilon_1 + 2} (V_1 + \beta)$$
(6)

 $(v_1 = volume of the solvent; E_1 = dielectric constant of the solvent)$

Similarly the electron polarisation may be expressed as:

$$P_{E_{2}} = \frac{6 \delta n_{1} V_{1}}{(n_{1}^{2} + 2)^{2}} + \frac{n_{1}^{2} - 1}{n_{1}^{2} + 2} (V_{1} + \beta)$$
(7)

(where $n = n_1 + \chi \omega_2$; $n_1 = refractive index of the solvent)$

By plotting $\boldsymbol{\varepsilon}$, v and n against $\boldsymbol{\omega}_2$; $\boldsymbol{\alpha}$, $\boldsymbol{\beta}$ and $\boldsymbol{\delta}$ may be obtained and by substituting these values into (6) and (7) the total polarisation, and electron polarisation may be evaluated.

At low concentrations ($w_2 \ll 0.02$) the graphs are linear¹²¹ and linear plots of n against w_2 may be obtained at even greater concentrations.

Apparatus and Experimental Details

The apparatus used is that previously described and used by P. $Dixon^{122}$ and J. G. Livingstone¹²³ with several minor modifications, and therefore is described somewhat briefly. All measurements were made at 25°C in benzene.

1. Dielectric Constant Measurements

The change in capacitance which occurred when the dielectric was changed of a capacitor (the dielectric cell) was measured. The cell of the Sayce-Briscoe type¹²⁴ with plates of platinum burnt on to the glass, could be filled with benzene or the benzene solution of the compound and solutions changed easily by blowing out with nitrogen, washing with pure benzene and replacing with the required liquid. It was immersed in a thermostat filled with transformer oil to reduce stray capacitances.

A heterodyne beat capacitance meter of the type described by Hill and Sutton¹²⁵ was used to measure the capacitance change in the cell. The X plates of a cathode ray tube carried

the output from a beat frequency generator which consisted of two oscillators, one of fixed frequency, f_0 (C.10⁵ c.p.s.), and the other of variable frequency, f_1 incorporating the measuring system. The circuits of the oscillators were as nearly identical as possible. The Y plates carried the output from an audiofrequency oscillator, f_2 (C. 1000 c.p.s.). The measuring system, shown below, consisted of the cell C, two air-spaced uncalibrated variable capacitors P and Q, and a calibrated concentric cylinder variable capacitor, M, with a micrometer screw, having a capacitance change of 3.38 pf. per cm. length. The cell and the calibrated capacitor could be interchanged by the switch, S.



P is adjusted until the output of the beat frequency generator gave a figure of eight on the cathode ray tube corresponding to $f_1 - f_0 = f_2/2$ which was used as a balance point throughout all the measurements. After filling the cell and allowing it to reach thermal equilibrium (about 10 minutes) P was adjusted. M was then switched in, to replace the cell, and adjusted in order to reobtain the figure of eight. After

adjusting each circuit several times, and when a steady figure of eight was obtained on both sides of the switch, the value of M was noted (arranging that for pure benzene the balance occurred with the calibrated capacitor reading about 2.4 cm., since its range was 0-25 cm.) After blowing out the benzene, the cell was washed out (twice with benzene) and then filled with a solution of the compound and the balance point again obtained. The difference in M, gave the difference in capacitance between the solution and pure benzene. Readings of M, with pure benzene in the cell, were taken after each solution, and an average value taken if a slight drift occurred.

If $\boldsymbol{\xi}_3$ is the dielectric constant of the solution and C_1 , C_2 and C_3 are the capacitances of the cell filled with benzene, air and solution respectively, then

$$\frac{c_3 - c_2}{c_1 - c_2} = \frac{\varepsilon_3 - \varepsilon_2}{\varepsilon_1 - \varepsilon_2}$$

$$\therefore \quad \varepsilon_3 = \left(\frac{c_3 - c_1}{c_1 - c_2} + 1\right)(\varepsilon_1 - \varepsilon_2) + \varepsilon_1$$

The dielectric constants of air (ξ_2) and benzene¹²⁶ (ξ_1) were taken as 1.0006 and 2.2727 respectively; $C_1 - C_2$ was taken as 69.05 pfs., as previously found¹²². $C_3 - C_1$ corresponds

to the change in capacitance (ΔC) measured with M.

$$\therefore \quad \xi_3 = \left(\frac{\Delta c}{69.05} + 1\right) 1.2721 + 1.0006$$

If l is the reading on the condenser in cm., as the capacitance change per cm. is 3.38 pfs., then,

$$\epsilon_3 = \left(\frac{3.38 \Delta \ell}{69.05} + 1\right) 1.2721 + 1.0006$$

$$\delta \mathbf{E}_{3} = \left(\frac{3 \cdot 38 \, \mathrm{Sl}}{69 \cdot 05}\right) 1 \cdot 2721$$

$$\frac{d\mathbf{E}}{d\mathbf{w}} = \frac{1.2721 \cdot 3.38}{69.05} \cdot \frac{d\mathbf{e}}{d\mathbf{w}}$$

$$= 0.06227 \cdot \frac{d\ell}{dw}$$

 $\frac{d\varrho}{dw}$ is the slope of the graph of (change in length on the calibrated condenser) against weight fraction and $\frac{d\varrho}{dw}$ corresponds to a in equation (6).

2. <u>Measurement of Refractive Index</u>

A Pulfrich refractometer with a divided cell, attached to the prism with silicone rubber, was used. The cell could be washed out with bromine in sodium bromide solution to remove

any gold which was deposited as a purple film by some of the less stable compounds. The refractometer was enclosed in a box fitted with a glove and window which could be lifted on a hinge. When closed it rested on polyurethane foam, to ensure a good fit and so avoid draughts. The temperature was kept at 25 \pm 0.2°C, throughout the box, by means of a fan, and heater in circuit with a thermostatted thermometer. A sodium lamp was generally used as a light source, but occasionally a projection lamp together with a red Jena Interference Filter $(\lambda_{max} = 6620^{\circ} A)$ was used. Pure benzene was placed in one side of the cell and a solution in the other, a polythene cap being placed over the cell to prevent evaporation. The window was closed and the solutions left for 15 minutes to attain equil-The difference in extinction angles between the solution ibrium. and pure solvent was measured.

The relationship between extinction angle and refractive index is given by $n = \sin A \sqrt{N^2 - \sin^2 B} + \cos A \sin B$ where, A is the refractive angle of the prism, B is the angle of emergence, and N is the refractive **matrix** of the prism. A graph of extinction angle against refractive index is linear and shows that 1 minute of arc = 1087 x 10⁻⁷ (λ = 7090 - 5893A^o). A graph of Δn against w_2 gives $\frac{dn}{dw} = \delta$ in equation (7).

3. Measurement of Specific Volume

The Sprengel type pykmometer used previously^{122,123} was

replaced by a similar one of larger capacity, as the accuracy of this measurement contributes appreciably to the final accuracy of the dipole moments. It was fitted with ground glass caps to prevent evaporation, and calibrated with distilled water. (Volume = 12.1208 cc.). Density measurements on pure benzene gave a value of 0.87376 gm./cc. (Literature gives 0.87368 gm/cc.¹²⁷). A solution in the pyknometer was placed in a water bath at 25° for 30 minutes and then on the balance for 10 minutes, to allow thermal equilibrium to be attained in each case.

Specific	volume	of	solution	12.1208
				weight of solution in
				the pyknometer

$$v = b + \beta w_2 = \frac{12.1208}{w_s}$$

$$\therefore \qquad \beta = \frac{12.1208}{w_s}$$

$$\frac{12.1208}{w_s}$$

 $oldsymbol{eta}$ is the slope of the graph of specific volume against weight fraction.

Example of Dipole Moment calculation from experimental results Compound: $C_6H_5C:CAuCNC_6H_4O-C_2H_5$ (phenylethynyl <u>o</u>-ethylphenylisocyanidegold) M = 429.3

	(1)	(2)	(3)	(4)
Wt. flask	43.9142	39•3344	49.6747	44.5370
Wt. flask and compound	43.9760	39.5549	50.0120	44.7668
Wt. flask and cpd. and benzene	60.7450	67.9070	79.7700	60.5920
Wt. compound	0.0618	0.2205	0.3373	0.2298
Wt. compound and benzene	16.7690	28.3521	29.7580	15.8252
Weight fraction $\omega \ge 10^6$	3685	7777	11355	14521
Dielectric_measurements				
Reading for pure benzene	2.489	2.485	2.485	
Reading for solution	1.626	0.825	0.100	
$\Delta c(cm.)$	0.863	1.660	2.385	
<u>Refractivity_measurements</u>				
$\Delta_n (\lambda = 5893A)$		9".00"	12'.40"	16'.10"
10^7 . Δ n		9783	13769	17573
Density measurements (wt. pyk	mometer	= 20.6429)	
Wt. pyknometer + sol ⁿ .		31.2776	31.2975	31.3144
. wt. sol ⁿ .		10.6347	10.6546	10.6715
Specific Volume V $\begin{pmatrix} = \frac{12.1208}{wt. sol^{n}} \end{pmatrix}$		1.13974	1.13761	1.13581
Graphs of AC , $10^6 An$, and V ag	ainst W	give:		
· · · · · · · · · · · · · · · · · · ·	alla			

 $dl = 0.06227 \times \frac{dllc}{dw}$ = 0.06227 x 209.8 = 13.0642

57.
$\beta = -0.5850$ $\delta = 0.1205$

From equation (6):

$$\frac{\text{Total Polarisation}}{\Gamma(\xi_{1}+2)^{2}} = \frac{3aV_{1}}{(\xi_{1}+2)^{2}} + \frac{\xi_{1}-1}{\xi_{1}+2} (V_{1}+\beta)$$

$$V_{1} = \text{specific volume of benzene} = 1.1445 \quad \xi_{1} = 2.2727$$

$$\therefore \text{ T.P.} = [0.1881 \quad \alpha + 0.2979 \quad (1.1445 + \beta)]M$$

$$\therefore \text{ T.P.} = [2.4574 + 0.1667]429.3$$

$$= 1126.5 \text{ cc.}$$

From equation (7): <u>Electron Polarisation</u> = $\frac{6n_1 \aleph v_1}{(n_1^2 + 2)^2} + \frac{n_1^2 - 1}{n_1^2 + 2} (v_1 + \beta)$

 $V_{1} = 1.1445, n_{1} = 1.4979 \text{ (for sodium D line)}$ $\therefore P_{E} = [0.5712\% + 0.2931(1.1445 + \beta)]M$ $\therefore P_{E} = [0.6883 + 0.16399]429.3$ = 99.9 cc.

 $P_0 = T_{\bullet}P_{\bullet} - (P_E + P_A)$

Assuming the atom polarisation to be 10 cc.

$$P_{o} = 1126.5 - 99.9 - 10$$

= 1016.6 cc.
.. Dipole moment, $\mu = 0.2212 \int P_{o}$
= 0.2212 J1016.6
.. $\mu = 7.05 D$









Accuracy of the measurements

For moments greater than 3.5D, the error is not greater than 0.1D. Moments of 1D have an error not greater than 20% (i.e. 0.2D), the error increasing rapidly as the moment approaches zero.

C. INFRA RED SPECTRA

Infra red spectra of solid compounds were generally taken in KBr at a concentration of 2-3mg. A Grubb Parsons GS2A spectrometer was used. The acetylenic frequencies given throughout this thesis are measured accurately and are correct to 1 cm^{-1} for frequencies above 2000 cm⁻¹, and to 2 cm⁻¹ for frequencies less than 2000 cm⁻¹. To measure an absorption frequency accurately, the approximate position was noted in the course of a run through the spectrum. The wavelength drum of the spectrometer was then adjusted to about 0.02 microns before the wavelength being measured. The recording chart was run 2 squares (i.e. 0.2" on the paper) at that wavelength. Then the wavelength drum was turned manually by one small division (corresponding to 0.001µ in the second order) and the chart run a further two squares. The wavelength drum was now moved a further division. and the chart run two more squares, and this process repeated until about 0.2µ after the peak which had now been traced out as a broader curve as shown in Figure 3. The maximum point on this curve corresponds to the accurate acetylenic stretching

frequency.

The acetylenic frequencies are collected in tables in the Discussion.

D. QUANTITATIVE ANALYSES

Micro carbon and hydrogen analyses are by Miss V. Conway and Mr. T. Caygill of this department. Other analyses were performed, in duplicate, as described.

Copper

(a) <u>Gravimetrically as copper(I)</u> thiocyanate.¹²⁸

About 0.1g., of the compound was decomposed by adding 5N nitric acid (2 cc.), warming gently and then adding conc. nitric acid (5 cc.) and boiling until the solution was pale blue. Concentrated hydrochloric acid (10 cc.) was added and boiling continued until no more oxides of nitrogen were evolved. The solution was cooled, diluted to 150 cc., and if a precipitate formed it was filtered off. A saturated solution of sulphur dioxide (about 20 cc.) was added to the filtrate and the solution boiled for 30 minutes. If the smell of sulphur dioxide persisted the solution was allowed to cool, if not more sulphur dioxide was added and boiling continued. The solution was made just acid and 2% ammonium thiocyanate solution (saturated with SO₂) added (10ml. for every 50mg of copper present). The precipitate was allowed to settle overnight, filtered through a sintered glass crucible (porosity 4) and dried at 110 - 120° for 3 hours.

(b) <u>Volumetrically</u>¹²⁹

When several copper analyses on compounds of uncertain were required constitution or purity, the more rapid volumetric estimation was carried out, decomposing the compound as described above, and estimating the iodine released on addition of potassium iodide to the copper solution, with standard sodium thiosulphate solution.

Gold

(a) <u>Compounds easily combusted</u> to gold (i.e. compounds containing only, C, H, N, Au, and sometimes P).

50-80mg. of the compound was weighed in a silica crucible with lid, which had previously been heated, cooled in a desiccator and weighed. The crucible and contents were heated with a **meeon** burner, very gently at first until the compound had decomposed, and then strongly until all organic matter had been combusted and the crucible containing only the gold residue had reached constant weight.

(b) Other compounds

About 0.1g. of the compound, was heated in a silica crucible until all organic matter had combusted. The crucible was allowed to cool, placed in a beaker (400 cc.) and the gold dissolved by addition of conc. nitric acid (5 cc.) and conc. hydrochloric acid (15cc.). The contents of the beaker were heated to remove oxides of nitrogen. The crucible was washed with water and

removed from the beaker, and the solution was diluted to 200cc. Any precipitate at this stage was filtered off. A solution of hydrozinehydrochloride was added to precipitate metallic gold, and the solution was boiled to coagulate the precipitate as much as possible. It was then filtered and the gold collected on a sintered disc (porosity 4) and dried at 150-170° to constant weight.

It is often difficult to transfer the last traces of gold from the beaker to the sintered disc, as when it is finely divided it sticks to the glass and policeman; analyses by this method, therefore, are usually slightly lower than the required value. (e.g. If 50% Au, is required; a value of about 49.5% would be acceptable).

<u>Nitrogen</u> was determined by the Kjeldahl method.¹³² <u>Mercury</u> was estimated gravimetrically as mercures sulphide.¹³⁰ <u>Silver</u> was estimated gravimetrically as silver chloride.¹³¹

CHAPTER III

COPPER, SILVER AND MERCURY ACETYLIDES

The copper, silver and mercury derivatives of various acetylenes were prepared by the well-known methods, and have all been reported previously. The preparations of the phenylacetylene derivatives are described in each case, and any interesting or unusual behaviour of other acetylenes is added. <u>Preparation and solubility properties of copper acetylides</u> (RC:CCu),

All yields are quantitative with the exception of t-butylethynylcopper. All the copper derivatives described, except t-butylethynylcopper, are insoluble in common organic solvents. They are all yellow, excepting <u>p</u>-nitrophenylethynylcopper and t-butylethynylcopper which are red, (though the latter is yellow if isolated by cooling a solution in ether).⁹²

Phenylethynylcopper

Copper(I) chloride (23g., 0.23 mole) was placed, in a nitrogen atmosphere, in a 500 cc., 3 necked flask equipped with stirrer and dropping funnel. Water (200cc.) was added, and sufficient concentrated ammonia to give a pale blue solution. Phenylacetylene (23g., 24.8cc., 0.2 mole) in ethanol (100cc.) was added with good stirring, an immediate bright yellow precipitate being formed. After addition was complete the solution was stirred for a further 30 minutes and the precipitate filtered. It was washed with 4N ammonia and water until the washings were colourless, and then with alcohol, acetone and ether and dried in vacuo (3 hours) with a sintered disc placed between the solid and the pump, so that the dry fine powder was not sucked into the vacuum system. Yield of phenylethynylcopper, 37g., 97%.

<u>Prop-l-yinyl</u>, <u>but-l-ynyl</u>, <u>pent-l-ynyl</u>, and <u>isopropenylethynyl</u> <u>copper</u> were prepared similarly with a condenser (acetone/CO₂, or water cooled) above the flask to avoid any loss of the acetylene. They were insoluble in amines as well as organic solvents.

The solid aromatic acetylenes were added as solutions in 1:2 dimethoxyethane.

<u>t-Butylethynylcopper</u>. On addition of t-butylacetylene to ammoniacal copper(I) chloride, a red granular precipitate was formed. After filtration, and washing with 4N ammonia and water, the red solid was dissolved in benzene. Any insoluble material was separated, the benzene removed and the t-butylethynylcopper recrystallised from hexane. (m.p. 148-150° dec.). The best yield obtained was only 55%. Ebullioscopic molecular weight measurements in benzene showed t-butylethynylcopper to be octameric. Found: M, 1140, 1161, 1177 in 0.73, 1.48, 2.95% benzene solution. $(C_6H_9Cu)_8$ requires M, 1157. The attempted preparation of "crystalline" phenylethynylcopper.

1. From copper(II)phenylpropiolate (PhC:CCO2)2Cu

Preparation of the copper salt. Phenylpropiolic acid (log., 68.4 m_mole) was dissolved in ethanol (20cc.) and added to sodium acetate (9.4g., 70 m_mole) in water (20cc.). A solution of copper(II) chloride (9.4g., 70 m_mole) in water (20cc.) was then added. On standing blue leaflets separated which were filtered off, washed with water and pumped dry. Yield 11.9g., 95%. An I.R spectrum showed SC:C at 2232 cm⁻¹. Decomposition to phenylethynylcopper_. Attempts were made to repeat the observation made by F. Straus¹³³ that on gently warming copper phenylpropiolate with various amines fine yellow needles of phenylethynylcopper are formed.

 $2(PhC:CCO_2)_2Cu \rightarrow PhC:C-C:CPh + 2PhC:CCu + 4CO_2$

Samples of the copper salt (lg.) were added to pyridine, quinoline, and isopropylamine (lOcc.) forming deep blue solutions. Slight warming caused evolution of CO_2 and the solutions turned greenish yellow with deposition of phenylethynylcopper. Even with very slow warming, or on keeping the solutions for a week at 25°C., the best specimen of phenylethynylcopper was only obtained as very fine feathery needles and was that from the pyridine solution.

2. Extraction of phenylethynylcopper with amines

Samples of phenylethynylcopper (2 - 3g.) were placed in Soxhlet extraction thimbles and extracted with various amines, $(N_2 \text{ atmosphere})$. From the isopropylamine and diethylamine extractions, microcrystalline phenylethynylcopper was obtained. Solutions containing a suspension of a few milligrams of the microcrystalline material in their respective solvents were kept in sealed tubes for several months, but no larger crystals grew.

Phenylethynylcopper could not be extracted by triethylamine and it is unaffected by 2,2-bipyridyl.

Preparation and solubility properties of silver acetylides (RC:CAg)_x

All yieldsof silver acetylides were quantitative.

<u>Phenylethynylsilver</u>. Silver nitrate (37.5g., 0.22 mole) was dissolved in a mixture of water (lOOcc.) and methanol (lOOcc.) in a 500cc. 3 necked flask. Concentrated ammonia was added until the precipitated silver oxide just redissolved. Phenylacetylene (20g., 0.20 mole) in methanol (30cc.) was added with good stirring. The white precipitate was washed thoroughly with water and then with methanol, acetone and ether, and dried in vacuo (2 hours). Yield 40g., 0.202 mole, 100%. An infra-red spectrum of the solid was identical with that of phenylethynylcopper except for the \Im CiC stretching, indicating that no

appreciable amount of silver nitrate was present.

<u>p-nitrophenylethynylsilver</u>, which was pale yellow, <u>p-brom-</u> <u>ethynylsilver</u> and <u>mesitylethynylsilver</u> were prepared similarly, adding the acetylene in 1,2-dimethoxyethane solution, and they are insoluble in common organic solvents. <u>p-nitrophenyl-</u> ethynylsilver could not be caused to explode as suggested in Beilstein.¹³⁴

But-l-ynyl, pent-l-ynyl and isopropenyl, prepared similarly, are soluble in cold chloroform, carbon tetrachloride, acetone and warm benzene with discolouration, and are insoluble in hexane and ether. Pentynylsilver is soluble in cold pyridine and in hot isopropylamine.

<u>t-Butylethynylsilver</u> is slightly soluble in boiling carbon tetrachloride, benzene and chloroform. Soxhlet extraction with benzene and cyclohexane afforded greyish feathery needles. It is insoluble in ammonia and triethylamine but soluble in cold pyridine and isopropylamine, extraction with the latter solvent giving colourless feathery needles.

Alkyl and Aryl ethynyl mercury(II)compounds (RC:C), Ha

These were all prepared by adding the acetylene in alcohol or 1,2-dimethoxyethane in the case of VI and VII (see table I) to alkaline potassium mercuri-iodide e.g. <u>bisphenylethynylmercury</u>.

Phenylacetylene (4g., 0.04 mole) in alcohol (10cc.), was added to alkaline potassium mercuri-iodide [mercuric chloride (11.1g., 0.041 mole), potassium iodide (27.5g.,

0.165 mole), sodium hydroxide (5g.) in water (200cc.)]. After stirring well for 30 minutes, the white precipitate was filtered, washed with alcohol, and pumped dry. Recrystallisation from methanol afforded bisphenylethynylmercury 5.5g., 92%.

Table 1 lists the mercury derivatives prepared, their solvents of recrystallisation and their melting points. The melting points were compared with those previously reported.

Table 1

Melting Points and Solvents of Recrystallisation for the compounds (RC:C)_Hg

	R	Solvent of Recrystallisation	Observed m.p.(°C)	Previously reported m.p.
I	Me	benzene	203-204	203-204 135
II	Et	ethanol	162-163	162-163 ¹³⁵
III	Pr^n	ethanol	119-120	117.9-118.3 136
IV	Bu ^t	ethanol	92-93	91-92 135
V	Ph	ethanol	125-126	124.5-125 135
VI	p NO ₂ C ₆ H	extraction with 4 acetone	dec. % 260	272-274 137
VII	p BrC ₆ H ₄	benzene	255-256(dec.)	256-256.5 138
VIII	mesityl- Me ₃ C ₆ H ₃	propanol	241-242(dec.)	238-239 136
IX	PhC:C-	propanol	194-195(dec.)	188-189 139

<u>Bisphenylbutadiynylmercury</u>. A complete analysis was made on this compound as it was at first thought that it was a new compound, and also to characterise the rather unstable parent

acetylene. (Found: C, 52.3; H, 2.5; Hg; 44.3, 44.8. Calc. for C₁₈H₂₀Hg, C, 53.3; H, 2.2; Hg, 44.5%).

CHAPTER IV

70.

COPPER ETHYNYL COMPLEXES WITH TERTIARY PHOSPHINES

Apparatus and Techniques

Preparations of copper (and silver) ethynyl complexes were carried out in a nitrogen atmosphere, with the exception of the triphenylphosphine complexes. Reactions were generally carried out in a double Schlenk tube¹⁴⁰ (Fig. 4 - lower half) with a sintered disc of porosity 3, placed between the two limbs. The reactants were added to and mixed in one limb, any impurities were then filtered off and the compound recrystallised in the other limb, collecting the product on the disc. Occasionally the disc would become blocked with insoluble particles. In such a case, or when it was necessary to transfer a solution to another Schlenk, this could be carried out easily using a Schlenk adapter placed above one limb of the Schlenk as shown in the upper half of the diagram. The solution or suspension was poured through a polythene tube on to the sintered disc B and filtered into the clean Schlenk. In this way a second filtration could be carried out under nitrogen simultaneous with the transference from one Schlenk to another. Crystallisations were often carried out below room temperature by clamping the limb of the Schlenk containing the solution inside the upper half of a dewar half filled with acetone/CO2



at -50° to -78°. The solutions were filtered through the disc while still cold, in order to separate the crystallised product.

Good yields were sometimes sacrificed to ensure a compound was as pure as possible, and so yields are not always given, though they were seldom less than 50%. Melting points are corrected.

Monophosphine-ethynylcopper complexes

Isopropenylethynyl(trimethylphosphine)copper CH2:CMeC:CCuPMe3

Isopropenylethynylcopper (4g., 0.031 mole) was suspended in benzene (20cc.) and trimethylphosphine (2.5g., 3cc., 0.034 mole) added. Removal of the benzene, and addition of hexane produced a yellow solution with some brown insoluble oily material which was filtered off. The clear filtrate was cooled to -50° C over acetone/CO₂, affording the crystalline complex (3.5g., 55%), m.p. 82-82.5, (Found: C, 46.8, H 6.9; Cu, 30.8, 30.9. C₈H₁₄CuP requires C, 47.0; H, 6.9; Cu, 31.0%). <u>t-Butylethynyl(trimethylphosphine)copper</u> Bu^tC:CCuPMe₃

t-Butylethynylcopper (5g., 0.035 mole) was suspended in benzene (30cc.). Addition of trimethylphosphine (2.8g., 3.5cc., 0.037 mole) changed the colour of the solution from red to yellow-orange. Removal of benzene and recrystallisation from hexane gave a pale yellow-green solid, m.p. 105-106°. (Found: C, 48.9; H 8.45; Cu 28.8. C₉H₁₈CuP requires C, 49.0; H 8.2; Cu 28.8%).

Phenylethynyl(trimethylphosphine)copper PhC:CCuPMe,

Phenylethynylcopper (lOg., 0.061 mole) was suspended in benzene (30cc.) and trimethylphosphine (4.8g., 6cc., 0.063 mole) was added, shaking well until all the acetylide had dissolved. Removal of the benzene and recrystallising the yellow solid from hot toluene gave yellow needle clusters, with a greenish fluorescence, (l2g., 69%), m.p. l16-l17⁰ (Found: C, 54.8; H, 6.1; Cu, 261. $C_{11}H_{14}$ CuP requires C, 54.9; H, 5.9; Cu 26.4%). Phenylethynyl(trimethylphosphine)copper is soluble in carbon tetrachloride, benzene and acetone, and is insoluble in hexane. <u>Phenylethynyl(triethylphosphine)copper</u> PhC:CCuPEt₃

This complex, reported previously 91,92 was prepared as above and crystallised from hexane as yellow needles. (Found: C, 59.6; H, 7.3. Calc. for $C_{14}H_{20}CuP$ C, 59.5; H, 7.1%). Phenylethynyl(triphenylphosphine)copper PhC:CCuPPh₃

Phenylethynylcopper (4.8g., 0.03 mole) was suspended in benzene (20cc.) and triphenylphosphine (8.0g., 0.031 mole) added. The suspension was kept at 70° for 30 minutes, and unreacted phenylethynylcopper (2g.) filtered off. On cooling, the solution deposited a very pale yellow-green solid (7g.). This solid became paler on standing, and on dissolving in hot benzene and leaving the solution to cool, a bright lemon coloured amorphous solid was deposited m.p. 212-214 (decomp.),

which would not redissolve in hot benzene, and was almost completely insoluble in all organic solvents. (Found: C, 73.2; H, 6.8; Cu, 14.4, 14.8. C₂₆H₂₀CuP requires C, 73.2; H, 4.7; Cu, 14.9%). I.R. spectra of this complex, and the pale coloured solid were identical.

p-nitrophenylethynyl(triphenylphosphine)copper p-NO2C6H4C:CCuPPh3

<u>p</u>-nitrophenylethynylcopper (2g., 0.01 mole) was added to triphenylphosphine (5.2g., 0.02 mole) in benzene (20cc.). On warming and shaking, all the copper acetylide dissolved. Addition of hexane precipitated orange crystalline needles (3g.), m.p. $222-224^{\circ}$, which were filtered off and dried. (Found: C, 66.0; H, 4.3; Cu, 13.7. C₂₉H₁₉CuNO₂P requires C, 66.2; H, 4.1; Cu, 13.5%). When these crystals were stood in contact with ethanol, in which they were insoluble, they turned into a bright red powdery material which had the same analysis and m.p., (Found: C, 66.2; H, 4.5; Cu, 13.7%), as the orange crystals. This red form dissolved in benzene to give an orange solution from which the orange crystalline form was recovered. Similarly the orange form gave a red solution in nitrobenzene.

The attempted preparations of some other monophosphine complexes Propynyl(trimethylphosphine)copper

Trimethylphosphine (3.2g., 4cc., 0.04 mole) was added to propynylcopper (4g., 0.038 mole) suspended in benzene (20cc.). Warming of the solution was avoided as decomposition was encouraged.

After shaking for 10 minutes, the solution was filtered and benzene removed from the filtrate. The resulting solid was dissolved in hexane, and on cooling to -50° , a white powder was precipitated. This white solid is the very unstable propynyl(trimethylphosphine)copper. It decomposed immediately on exposure to air, and more slowly in a nitrogen atmosphere. <u>Addition of dimethylphenylphosphine to propynylcopper</u> in benzene, afforded a red viscous oil, which solidified at low temperatures and was soluble in hexane. On exposure to air it turned black.

Propynylcopper and triphenylphosphine did not react on boiling a suspension in benzene solution.

Reaction between isopropenylethynylcopper and dimethylphenylphosphine.

Dimethylphenylphosphine (4.25g., 4.4cc., 0.024 mole) was added to a benzene (20cc.) suspension of isopropenylethynylcopper (3g., 0.023 mole). Removal of the benzene afforded a red oil containing fine red crystals. Some of the oil dissolved on addition of acetone, the rest (0.5g.) remaining as a red powder, m.p. 135-138° (decomp.). Attempts to recrystallise this solid from various solvents were unsuccessful.

Reaction between t-butylethynylcopper and triphenylphosphine

On addition of triphenylphosphine (3g., 0.0115 mole) in benzene (25cc.) to t-butylethynylcopper (1.5g., 0.0104 mole)

in benzene (20cc.) the solution immediately changed from red to yellow. The benzene was pumped off until about 8cc. remained, and the solid which had been precipitated was filtered (4g., m.p. 115-118°). On recrystallising from hexane it had m.p. 145-148°, (Found: C, 70.4; H, 6.6; Cu, 13.3, 14.3. T-butylethynyl(triphenylphosphine) copper, C₂₄H₂₄CuP, requires C, 70.8; H, 5.95; Cu, 15.6**g%**). Complexes with triphenylphosphine are difficult to separate from excess triphenylphosphine, because of similar solubility properties. The required complex was probably obtained, but was contaminated with a little triphenylphosphine, which would account for the observed m.p. and analysis.

Reaction between phenylethynylcopper and trim-tolylphosphine

Phenylethynylcopper (1.5g., 9 m mole) was suspended in benzene (25cc.) containing trimtotylphosphine (2.6g., 8.6 m mole). The suspension was warmed, shaking vigorously, and after 1 hour phenylethynylcopper (1.2g.) was filtered off. Evaporation of the filtrate gave a brown oil which afforded a pale yellow solid (0.4g., m.p. 125-127), on addition of ethanol. This solid dissolved in hot propanol, and on cooling a bright yellow precipitate, which was soluble in benzene, was formed. An I.R. spectra of this yellow solid indicated that reaction had taken place to a slight extent, forming some phenylethynyl-(trim-tolylphosphine).

<u>Reaction between mesitylethynylcopper and triethylphosphine</u> afforded a yellow complex, recrystallised from hexane, but no satisfactory analysis was obtained though it seemed certain that a l:l complex was formed as one equivalent of the phosphine was sufficient to dissolve an equivalent of mesitylethynylcopper suspended in benzene.

Reaction between p-bromophenylethynylcopper and triethylphosphine

p-bromophenylethynylcopper (5g., 0.0204 mole) was suspended in benzene (30cc.) and triethylphosphine (2.49g., 3.12cc., 0.0210 mole) added. Only part of the acetylide dissolved. After filtration, the benzene was removed from the filtrate leaving a yellow solid (m.p. decomp. > 180°). Further triethylphosphine (0.0210 mole) was added to the unreacted acetylide which then afforded more of the yellow solid. These materials were combined in hot hexane, and on cooling the solution to 15° some small yellow crystals separated and were collected (0.8g.). Found Cu, 14.95, 15.2). Further cooling to -50° gave a yellow powder (4.lg.). (Found: Cu, 16.3, 16.4. BrC₆H₄C:CCuPEt₃ requires Cu, 17.6. BrC₆H₄C:CCu.2PEt₃ requires Cu, 13.2%.). This investigation was left at this stage to carry out other work. More than one complex was apparently obtained, and the predominate one contained P:Cu ratio > 1.

Reaction between phenylethynylcopper and 1,2-diphenylphosphineethane.

Phenylethynylcopper (4.1g., 0.025 mole) was suspended in

benzene (20cc.), and did not dissolve on addition of 1,2-diphenylphosphinoethane (5.0g., 0.0126 mole) in benzene (20cc.). The benzene was pumped off and replaced by chloroform and the suspension of phenylethynylcopper refluxed for 2 hours. The hot suspension was filtered and on allowing the filtrate to cool a pale cream powder, 2.8g., m.p. $300-304^{\circ}$, was deposited. This material on analysis gave: C, 52.4; H, 4.8; Cu, 8.95, 9.16%. A compound of formula (PhC:CCu)₂(Ph₂PCH₂CH₂PPh₂) requires a larger percentage of these elements. Its infra-red spectrum showed no absorption due to acetylenic stretching. Unreacted phenylethynylcopper (2.8g.) was also recovered from the reaction. <u>Complexes with P:Cu ratio ≥ 1 </u>

t-Butylethynyl(bistrimethylphosphine)copper Bu^tC:CCu.(PMe3)2

t-Butylethynylcopper (3g., 0.0208 mole) was suspended in benzene (25cc.). On addition of trimethylphosphine (3.36g., 4.2cc., 0.0432 mole) the solution became very pale green. Removal of benzene left a white solid contaminated with some brown impurity. The white solid was readily soluble in hexane, and was recrystallised (3 times) by cooling the hexane solution to -50° . Yield 3g., 48.5%, m.p. 95-98° (sealed tube) (Found: C, 46.4, 46.3; H, 9.1, 9.3; Cu, 21.2. $C_{12}H_{27}CuP_2$ requires C, 48.55; H, 9.2; Cu, 21.4%). The carbon analysis was not improved with further recrystallisations. The complex was more soluble than the monophosphine compound in similar solvents, and much more air-sensitive.

Phenylethynyl(bistrimethylphosphine)copper PhC:CCu(PMe3)2

Trimethylphosphine (3.8g., 4.8cc., 0.050 mole) was added to phenylethynylcopper (4g., 0.024 mole) suspended in benzene (25cc.). The solid obtained on removing the benzene, was crystallised from cyclohexane as pale yellow-green fine crystals (2.5g.) which deepened to a yellow-green solid (m.p. 94-96°) on pumping (45 minutes). (Found: C, 52.9; H, 7.0; Cu, 20.1, 20.0. $C_{14}H_{23}CuP_2$ requires C, 53.1; H, 7.3; Cu, 20.05%). Phenylethynyl(tristrimethylphosphine)copper PhC:CCu(PMe₃)₃

Phenylethynyl(trimethylphosphine)copper (2.2g., 0.0093 mole) was dissolved in benzene (25cc.) and trimethylphosphine (1.5g., 1.9cc., 0.02 mole) added to give a pale yellow solution. The benzene was removed and the solid crystallised from cyclohexane as small pale cream crystals which were collected on the sintered disc of the Schlenk tube. After pumping for a few seconds on a water pump, they were dried by blowing a stream of nitrogen through the disc. Yield (2.5g., 67.5%). (Found: C, 51.5; H, 9.4; Cu, 16.3, 16.4. $C_{17}H_{32}CuP_{3}$ requires C, 52.0; H, 8.2; Cu 16.2%). On pumping this solid it deepened slightly to pale yellow but only a negligible amount of volatile material was collected on a vacuum line.

<u>p-nitrophenylethynyl(bisdiethylphenylphosphine)copper</u> <u>p-NO₂C₆H₄C:CCu(PPhEt₂)₂.</u>

Diethylphenylphosphine (1.9g., 1.9cc., 0.011 mole) was

added to <u>p</u>-nitrophenylethynylcopper (2g., 0.01 mole) in benzene (20cc.) a red solution being formed immediately though part of the acetylide remained unreacted, but dissolved on addition of further phosphine (0.011 mole). After filtering off any impurity the benzene was removed until the volume was about locc.. Hexame (70cc.) was now added and the violet-red almost black crystals (1.5g., m.p. 75°) settled from the dark red solution, on standing for a few days. Attempts to recrystallise from acetone, methycyclohexane and various solvent combinations were unsuccessful. (Found: C, 61.6; H, 6.6; Cu, 11.7. $C_{28}H_{34}CuNO_2P_2$ requires C, 62.0; H, 6.3; Cu 11.7%).

Attempts to prepare a complex of <u>p</u>-nitrophenylethynylcopper with trimethylphosphine were unsuccessful. Again two equivalents of trimethylphosphine were required to dissolve one equivalent of <u>p</u>-nitrophenylethynylcopper forming an intense, dark-red solution. Some very dark red crystals (0.2g.) were obtained which contained: C, 49.7; H, 6.1. $C_{11}H_{13}CuNO_2P$ requires C, 46.1; H, 6.0. $C_{14}H_{22}CuNO_2P_2$ requires C, 59.55 H, 7.85%.

<u>Mixed ethynyl complexes</u> RC:CCu.R'C:CCu.2PMe₃ <u>t-butylethynyl(phenylethynyl)(bistrimethylphosphine)dicopper</u>

t-butylethynylcopper (2g., 0.0138 mole) was dissolved in benzene (30 cc.) and phenylethynylcopper (2.27g., 0.0138) added, followed by trimethylphosphine (0.0276 mole, 15cc. of a

standard solution in benzene) which produced an immediate colour change from red to yellow, with solution of all the phenylethynylcopper. Thebenzene was removed, and the solid recrystallised from benzene-hexane, as small yellow crystals. 3g. m.p. 105-111°. A second recrystallisation from toluenehexane raised the m.p. to 111-112°. (Found: C, 51.8; H, 7.0; Cu, 27.6, 27.7. C₂₀H₃₂Cu₂P₂ requires C, 52.05; H, 7.0; Cu 27.5%). An equimolar mixture of phenylethynyl(trimethylphosphine)copper and t-butylethynyl(trimethylphosphine)copper had m.p. 70-75 (dec.) indicating that a new compound and not the equimolar mixture had been isolated. One acetylenic stretching frequency at 2062 cm.⁻¹ (KBr disc) was observed in the I.R. The infra red spectra was otherwise identical with that of the equimolar mixture (\Im C=C 2058 cm⁻¹) except that a weak peak at 1898 cm⁻¹, a shoulder at 953 cm⁻¹, and a broad absorption at 500 cm⁻¹ were absent.

Reaction between methylethynylcopper, phenylethynylcopper and triethylphosphine

On addition of triethylphosphine (4.72g., 5.9cc., 0.04 mole) to propynylcopper (2.05g., 0.02 mole) and phenylethynylcopper (3.3g., 0.02 mole) suspended in benzene (30cc.), only the phenylethynylcopper dissolved, methylethynylcopper (2g.) being recovered on filtration. The benzene was removed, and the solid left, recrystallised from hexane gave yellow needles (2.6g.), presumably phenylethynyl(bistriethylphosphine)copper.

Reaction between phenylethynylcopper, isopropenylethynylcopper and trimethylphosphine

Isopropenylethynylcopper (2.5g., 0.02 mole) and phenylethynylcopper (3.3g., 0.02 mole) were suspended in benzene (30cc.), and all dissolved om addition of trimethylphosphine (0.04 mole, 26cc. of a standard solution in benzene). Removal of the benzene left an oily material, difficult to recrystallise, which gave from toluene-hexane solution, a yellow solid (2g.). (Found: Cu, 34.7; 36.4. The mixed complex requires Cu, 28.7; and the isopropenylethynyl and phenylethynyl(trimethylphosphine) complexes require Cu, 31.0, and 26.4% respectively). In view of the difficulty of isolating any pure compound, this reaction was left at this stage.

Phenylbutadiynylcomplexes with trialkylphosphines

Phenylbutadiynyl(trimethylphosphine)copper PhC:C'C:CCuPMe3

Phenylbutadiynylcopper (1.5g., 7.9 m mole) was suspended in benzene (20cc.) in which it dissolved on addition of trimethylphosphine (0.62g., 0.77cc., 8.0 m mole). After removing the benzene, and recrystallising from toluene, the complex was again recrystallised from benzene-methylcyclohexane, (m.p. decomp. > 140°). The pure complex may be recrystallised from benzene. (Found: C, 58.5; H, 5.5; Cu, 23.9, 23.6. $C_{13}H_{14}CuP$ requires C, 58.95; H, 5.3; Cu, 24.0%). Phenylbutadiynyl(triethylphosphine)copper PhC:C.C:CCuPEt3

Phenylbutadiynylcopper (3g., 0.023 mole) was suspended in benzene (30cc.) and triethylphosphine (2.96g., 3.7cc., 0.025 mole) added. The benzene was pumped off and the yellow-brown residue recrystallised from hexane (4 times) affording yellow needles, m.p. 155-156 (dec.). Recrystallisation may also be carried out from propanol. (Found: C, 62.3; H, 6.7; Cu, 20.5; 20.8. $C_{16}H_{20}$ CuP requires C, 62.6; H, 6.6; Cu, 20.7%). Bisphenylbutadiynyl(tristriethylphosphine)copper (PhC:C.C:CCu)₂ (PEt₃)₃

Addition of triethylphosphine (0.461g., 0.52cc., 3.5 m mole) to a suspension of phenylbutadiynyl(triethylphosphine)copper (1.1g., 3.5 m mole), in benzene (20cc.), which was then removed, afforded a yellow solid, soluble in warm hexane recrystallising as pale yellow transparent needles (1g., 78%, m.p. 95-96°). (Found: C, 62.4; H, 7.7; Cu, 17.4, 17.8. C₃₈H₅₅Cu₂P₃ requires C, 62.4; H, 7.6; Cu, 17.4%). A small amount was first isolated during a preparation of phenylbutadiynyl(trip(ethylphosphine)copper, using a slight excess of phosphine. This complex with Cu:P ratio 2:3 loses triethylphosphine on standing in air or on heating a solution in propanol, forming the monophosphine complex. <u>Phenylbutadiynyl(tri-n-propylphosphine)copper</u> PhC:C.CCuPPrⁿ₃

Addition of tripropylphosphine (2.8g., 3.5cc., 17.5 m mole) to a suspension of phenylbutadiynylcopper (3g., 15.9 m mole) in benzene (25cc.) followed by removal of the benzene and recrystallisation of the residue from propanol gave yellow needles (3g., 54%, m.p. 87°). (Found: C, 65.2; H, 7.8; Cu, 18.3,18.3. $C_{19}H_{24}CuP$ requires C, 65.4; H, 7.5; Cu 18.2%). Infra Red Spectra

Infra-red acetylenic stretching frequencies of the copper complexes are collected in Table 5, page 124. They were frequently measured in solution in benzene as well as in KBr discs. Carbon tetrachloride and tetrachlorethylene are very transparent in the acetylenic region, but unfortunately they slowly react with aliphatic tertiary phosphines and their complexes Benzene absorbs quite appreciably near 2000 cm.⁻¹ but (with slow scanning) was found to be the most suitable solvent.

Two series of complexes were studied in more detail, $RC:CCu(PMe_3)_n$, (R = phenyl, isopropenyl). Solutions containing Cu:P in the ratio, 1:1, 1:2, 1:3 and 1:4 were prepared by dissolving the mono phosphine complex (lOng.) in benzene (lcc.) and adding the appropriate amount of trimethylphosphine (as a standard solution in benzene). The samples were introduced, in turn, into a l mm KBr cell, a blank of pure benzene in a similar cell being placed in the reference beam of the spectrometer. Table **6**, page 125, shows the various frequencies. In the case of the isopropenyl series, addition of phosphine was accompanied by a lightening of colour of the solution, and when n = 3 and 4, the solutions were colourless. The t-butylethynyl series of complexes gave only very very weak acetylenic stretching frequencies. Extinction coefficients were calculated for the various absorptions.

SILVER ETHYNYL COMPLEXES

Phenylethynyl(trimethylphosphine)silver PhC:CAgPMe3

Trimethylphosphine (0.035 mole, 7 ml. of a solution in hexane) was added to phenylethynylsilver (3.96g., 0.02 mole) suspended in hexane (30cc.). After shaking for 20 minutes the hexane was pumped off and the complex recrystallised from a large volume of acetone as white needles, m.p. dec 130° , (Found: C, 46.5; H, 5.3; Ag, 37.7%. $C_{11}H_{14}AgP$ requires C, 46.3; H, 4.95; Ag, 37.9%). Phenylethynyl(trimethylphosphine)silver is very light sensitive, and should be prepared in subdued light and stored in the dark. In 2.07, 2.29, 3.82 wt.% benzene (ebullioscopically) it has an association of 2.21, 1.82, 2.00. <u>Reaction between t-butylethynylsilver and trimethylphosphine</u> afforded some fawn needles probably of the l:l complex, but no satisfactory analysis was obtained.

CHAPTER V

GOLD (I) ETHYNYL COMPLEXES

PHENYL ETHYNYL GOLD AND ITS COMPLEXES Preparation of phenylethynylgold (PhC:CAu)_x

The preparation of gold acetylides must be carried out carefully to avoid formation of metallic gold. Apuric(III) bromide is reduced with sulphur dioxide to the gold(I) state in an aqueous acetone medium, avoiding the presence of excess SO₂. In aqueous ethanol solution, gold was more easily formed.

An aqueous solution of gold(III)chloride $(HAuCl_4 \cdot xH_2O,$ 51% Au, lOg., 0.026g., atom gold) was filtered into a 500cc. flanged flask (of the type shown in Fig. 1, without the tap in the bottom of the flask) equipped with a 'vibromixer' stirrer and a gas lead ending just below the stirrer. A filtered aqueous solution of potassium bromide (15g., 0.115 mole) was added and the volume made up to 200cc. with acetone, so that the acetone and water were present in roughly equal volumes. A slow stream of SO₂ was passed through with vigorous stirring, until the solution turned pale red. The SO₂ was then passed very slowly and stopped when the solution became pale yellow. Phenylacetylene (2.6g., 2.78cc., 0.0255 mole) in acetone (10cc.) was added, followed immediately by excess sodium acetate (c. 25g. of the hydrated salt in water(100cc.). The initial pale yellow precipitate gradually thickened and after stirring for 30 minutes was filtered, washed with acetone and ether and dried at the pump. Yield of phenylethynylgold, 7g., 91%. (Found: C, 32.0; H, 1.8; Au, 65.2, 65.7. C_8H_5Au requires C, 32.2; H, 1.7%; Au, 66.1%). Phenylethynylgold on heating remains yellow to about 100°, but decomposes gradually above this temperature. It appears to possess no explosive property.

Complexes of phenylethynylgold

The complexes were generally prepared by suspending phenylethynylgold in benzene and adding a slight excess of the donor, followed by removal of the benzene and recrystallisation of the solid from the solvent stated. In most cases warming encouraged decomposition of the complex, with the formation of gold; they were, therefore, generally recrystallised by cooling a solution of the complex at room temperature, to -20° in a "deep-freeze" or to lower temperatures over an acetone/CO₂ bath. Reagents were added under nitrogen, but once the compound had been formed it was handled and recrystallised in air. The complexes must be stored in the cold, and more essentially in the dark. When pure, they are all colourless, and they were obtained in quantitative yields.

Phenylethynyl(ammine)gold PhC:CAuNH3

In the presence of ammonia, phenylethynylgold turns to a

colourless complex, insoluble in all common solvents. It is best prepared by adding an ammoniacal acetone solution to phenylethynyl(<u>n</u>-octylamine)gold, when it is precipitated as colourless crystals, dec. > 120° . (Found: C, 32.3; H, 2.8, Au, 62.5, 62.3. C₈H₈AuN requires C, 30.5; H, 2.6; Au, 62.5%). <u>Phenylethynyl(isopropylamine)gold</u> PhC:CAuNH₂Pr¹

Excess isopropylamine was added to a benzene (lOcc.) suspension of phenylethynylgold (l.lg.). The benzene and excess amine were removed and the complex crystallised from acetone as fine colourless needles (lg.). (Found: C, 36.7; H, 4.l; Au, 54.8, 55.l. $C_{11}H_{14}$ AuN requires C, 37.0; H, 3.95; Au, 55.2%). The complex darkens rapidly above 115° , is soluble in hot acetone and benzene, and insoluble in carbon tetrachloride, chloroform, ether and methylcyclohexane.

Phenylethynyl(n-pentylamine)gold PhC:CAuNH₂C₅^{**} H₁₁ⁿ

Similarly prepared to the above complex, this was crystallised as white feathery needles by cooling to -20° its solution in 1,2-dimethoxyethane. (Found: C, 40.9; H, 4.8; Au, 51.1, 51.0. $C_{12}H_{16}AuN$ requires C, 40.5; H, 4.7; Au 51.1%). It darkens above 115° and melts at 124-125° (decomp.). It is moderately soluble in warm acetone, chloroform, and carbon tetrachloride, but insoluble in ether and in hexane.

Phenylethynyl(n-octylamine)gold PhC:CAuNH2C8H17ⁿ

Similarly prepared, but more soluble in similar solvents,

this complex was crystallised as colourless needles from ,2-dimethoxymethane, m.p. dec. above 120 . (Found: C, 45.0; H, 5.6; Au, 45.7, 45.8. $C_{16}H_{24}AuN$ requires C, 44.9; H, 5.7; Au, 46.1%. Degree of association 3.08, 3.40, 3.79, 3.54, 4.12 cryoscopically in 1.34, 2.87, 3.39, 4.42, 5.73 wt. % benzene).

Phenylethynyl(n-nonylamine)gold PhC:CAuNH₂C₉H₁₉ⁿ

This complex crystallised as colourless needles from 1,2-dimethoxyethane, m.p. 117-118° (decomp., and darkening above 105°). (Found: C, 46.6; H, 5.9; Au, 44.6, 44.4. C₁₇ H₂₆AuN requires C, 46.2; H, 5.9; Au, 44.7%. Degree of association 2.58, 3.12, 3.30, 3.62, 4.06 cryoscopically in 0.73, 1.56, 2.62, 3.62, 5.15 wt. % benzene). Both this and the octylamine complex are not sufficiently soluble in nitrobenzene at its freezing point to carry out molecular weight measurements. <u>Phenylethynyl(di-n-butylamine)gold</u> PhC:CAuNH(C₄H₉ⁿ}₂

Crystallised from acetone as colourless needles m.p. $91-92^{\circ}$ (decomp.), a solution of this complex in benzene darkens rapidly and is unsuitable for dipole moment or molecular weight measurements. (Found: C, 45.5; H, 5.7; Au, 46.2, 46.3. $C_{16}^{H}_{24}$ AuN requires C, 40.7; H, 4.2; Au, 51.4%).

Phenylethynyl(pyridine)gold

Excess pyridine was added to phenylethynylgold suspended in l,2-dimethoxyethane. On cooling to -20° , fawn needles separated, decomp. above 80° . (Found: C, 42.0, 40.1; H, 3.0,
2.6; Au, 52.1, 52.3. $C_{13}H_{10}$ AuN requires C, 41.4; H, 2.7; Au, 52.25%). The pure complex lost pyridine when suspended in benzene and phenylethynylgold (identified by I.R. spectra and gold analysis) was recovered. Phenylethynyl(pyridine)gold is insoluble in most solvents, dissolving on addition of excess pyridine. Pyridine did not replace octalamine from a benzene solution of its phenylethynylgold complex.

Phenylethynylgold and some tertiary amines

Phenylethynylgold dissolved in a benzene solution containing an excess of triethylamine or tributylamine. The solutions darkened rapidly on standing and after $l\frac{1}{2}$ hours a black sludge separated. Phenylethynylgold does not react with bipyridyl.

Phenylethynyl(trimethylphosphine)gold PhC:CAuPMe3

This complex was crystallised, as colourless very fine powder from a toluene-hexane mixture, m.p. $190-191^{\circ}$ (decomp.). (Found: C, 33.9, H, 3.9. $C_{11}H_{14}$ AuP requires C, 35.3; H, 3.8%). It is insoluble in benzene and difficult to obtain (well defined crystals.

Phenylethynyl(triphenylphosphine)gold PhC: CAuPPh

This complex was crystallised as very pale cream plates from a benzene-methylcyclohexane mixture, m.p. 163-164^o (decomp.) (Found: C, 56.1; H, 3.8; Au, 35.1; M, cryoscopically in benzene 533, 482, 482. C₂₆H₂₀AuP requires C, 55.7; H, 3.6; Au, 35.2%, M; 561. It is soluble in acetone, benzene, chloroform, carbon tetrachloride, slightly soluble in ether, and insoluble in hexane.

Bisphenylethynyl(1,2-diphenylphosphino-ethane)digold (PhC:CAu)₂Ph₂PCH₂CH₂PPh₂.

1,2-diphenylphosphino-ethane (0.8g., 2 m mole) was added to phenylethynyl(n-octylamine)gold (1.7g., 4 m mole) in chloroform (20cc.). The solution was filtered to remove any impurity and ethanol added until a precipitate just began to form. On standing, # fine white needles (decomp. above 200°) separated and were filtered off. (Found: C, 49.9; H, 3.7; Au, 39.7. $C_{42}H_{34}Au_2P_2$ requires C, 50.6; H, 3.4; Au, 39.6%). It is insoluble in cold benzene, in contrast to the other phosphine and similar complexes.

Phenylethynyl(triphenylphosphite)gold PhC: CAuP(OPh)3

This complex was obtained as colourless feathery needles from ethanol, m.p. 113-114°. (Found: C, 50.8; H, 3.2; Au 32.3, 32.0; M, cryoscopically in benzene 548, 596, 578, 596. $C_{26}H_{20}$ AuO₃P requires C, 51.3; H, 3.3; Au, 32.4%; M, 609). Phenylethynylgold and trialkylphosphites

Phenylethynylgold readily dissolves in benzene solutions of trimethyl or triethylphosphite, but the exceedingly soluble complexes were obtained as oils and attempts to purify them were

unsuccessful.

Phenylethynyl(triethylarsine)gold PhC:CAuAsEt,

This complex was isolated both as pale yellow needles and as fine grey needles, in each case crystallising from ether cooled to -25° , m.p. 67-68°. (Found: C, 35.9; H, 4.7; Au, 42.7; M, cryoscopically in 0.46, 0.66, 1.01 wt. % benzene, 474, 484, 460. $C_{14}H_{20}AsAu$ requires C, 36.5; H, 4.6; Au, 42.8%; M, 460).

Phenylethynyl(triethylstibine)gold PhC:CAuSbEt

On addition of triethylstibine (1 mol.) to phenylethynylgold (3g.) a pale yellow solution was formed. Removal of the benzene left an oily material, difficult to recrystallise, only a small amount of fairly pure material being obtained from propanol. (0.2g., m.p. 96-97°. Found: C. 32.7; H. 3.2. C C14H20AuSb requires C, 33.1; H, 4.0%). A product with identical infra red spectrum was also obtained by displacement of n-octylamine, from its phenylethynylgold complex, by triethyl-It was recrystallised from methanol as pale off-white stibine. needles, m.p. 60-70°. (Found: C, 31.5; H, 4.5%). Phenyl(triethylstibine)gold is difficult to purify and is not very stable darkening rapidly in air though more slowly under nitrogen. It was handled under nitrogen in a Schlenck tube. Phenylethynyl(n-butylisocyanide)gold PhC:CAuCNBuⁿ. was crystallised from ether as colourless feathery needles.

m.p. 40.5-41.5°. (Found: C, 41.0; H, 3.8; Au, 51.5, 51.5; M, cryoscopically in 1.00, 1.50, 1.79, 1.99 wt. % benzene, 386, 409, 417, 412. $C_{13}H_{14}AuN$ requires C, 40.9; H, 3.7; Au, 51.7%; M, 382). It is soluble in chloroform, acetone, benzene and ethanol, and insoluble in hexane.

<u>Phenylethynyl(p-tolylisocyanide)gold</u> PhC:CAuCN $C_{6}H_{4}p-CH_{3}$ was isolated as colourless leaflets from benzene, in which it is only sparingly soluble in the cold, m.p. 176-177^o (decomp.). (Found: C, 46.2; H, 2.8; Au 47.7. $C_{16}H_{12}AuN$ requires C, 46.3; H, 2.9; Au 47.5%).

<u>Phenylethynyl(o-ethylphenylisocyanide)gold</u> PhC:CAuCNC₆H₄ $_{0}$ C₂H₅ was crystallised as colourless needles from 1,2-dimethoxyethane, m.p. 107-108°. (Found: C, 47.5; H, 3.4; Au, 45.7; 45.7; M, cryoscopically in 0.95, 1.43, 1.72, 1.91 wt. % benzene, 439, 400, 446, 427. C₁₇H₁₄AuN requires C, 47.55; H, 3.3; Au, 45.9%; M, 429).

With the exception of the amine complexes and that formed by 1,2-diphenylphosphino-ethane, most complexes were freely soluble in organic solvents except saturated hydrocarbons.

Dipole Moment Data

Atom Polarisations are assumed to be lOcc. for each gold atom in the molecule.

PhC:CAuPPh₃

wt. fraction 10^6 w	$\Delta_{c(cm)}$	Specific Volum V	$10^7 \Delta$ n
3882	0.561	1.1433	4529
6028	0.759	1.1410	7066
7642	1.032	1,1411	7790
10487	1.512	1.1386	11957
18784		1.1350	21200
$\alpha = \frac{\mathrm{d}\Delta c}{\mathrm{d}w} \mathbf{x}$	$0.06227 \beta = \frac{\mathrm{d}V}{\mathrm{d}w}$	= - 0.5507 \delta =	$\frac{\mathrm{dn}}{\mathrm{dw}} = 0.1132$

= 8.9631

Total	Polarisation		104	44 cc.					
Electr	on Polaricati	on	-	1 2 2 1	00.		±n ≍	= 6.7	T
	OH TOTSTISCOT	UTT .		±•رر⊥			<u>µ</u>		<u></u> •

PhC:CAuP(OPh)3

	10 ⁶ w	Δc	V		10 ⁷ Δ n
	4402	0.270	1.1422		4167
	6803	0.319	1.1411		6522
	8609	0.563	1.1402		7790
	15440	0.936	1.1364		14312
a	= 3.7854	$\beta = -0.5305$: :	8 =	0.09529

Total Polarisation = 544.6 cc.

Electron Polarisation = 140.9 cc.

u = 4.4 D.

94**a**.

PhC:CAuAsEt3			
l0 ⁶ w	Δc	V	lo ⁷ A n
4647	0.708	1.14343	4076
7220	1.087	1.14013	5616
10109	1.493	1.13857	7066
13334	2.069	1.13618	9421
19472		1.13212	13689
a = 9.4339	$\beta = -\alpha$	0.6450	X = 0.07025
Total Polarisa	ation = 885.2 co	с.	anta ana amin'ny fisiana amin'ny fisiana Ny INSEE dia mampina mampina amin'ny fisiana Ny INSEE dia mampina mampina mampina mampina mampina mampina mampina
Electron Polar	visation = 85.9	cc.	$\mu = 6.2 D$
PhC:CAuCNBn ⁿ			
lo ⁶ w	Δc	V	10 ⁷ (1n
37 39	0•965		
4690	1.060	1.14176	
6417	1.422	1.14026	5 7 97
8954	1.985		7247
10367		1.13853	
13324		1.13681	11051
19136			15940
α = 13.917	$\beta = -0.58$	380	X = 0.08300
Total Polarisa	tion = $1062 cc$.		

81.2 cc.

Electron Polarisation =

 $\mu = 6.9 D$

PhC:CAuCNC ₆ H ₄	<u>- C₂H₅ -</u>	See page 57	
PhC:CAUNH2C5H11	n		
10 ⁶ w	Δc	V	10 ⁷ 0 n
4354	0.058	1.14217	2174
7390	0.103	1.14010	4348
12850	0.161	1.13593	7609
17104	0.216	1.13391	10690
a = 0.7784	$\beta = 0.6290$	8 =	0.05925
PhC: CAUNH2C8H17	<u>n</u>		
10 ⁶ w	Δc	V	10 ⁷ / ₁
4468		1.14280	
8499		1.14039	4892
12796		1.13916	17971
16703		1.13604	10327
18281			11232
6719	0.109		
9168	0.130		
11595	0.205		
$\alpha = 0.7939$	$\beta = -0.1$	5175 🔏 =	= 0.06200

b.

PhC: CAUNH2C9H19

lo ⁶ w	Δ°	V	10 ⁷ (1 m
5381	0.046		
7365	0.099		
745 7		1.14085	4167
10679		1.13910	5797
12978	0.164		
14594		1.13717	7066
	-		

 $\alpha = 0.7845$ $\beta = -0.5050$ $\lambda = 0.0545$

From m.w. measurements in benzene the degree of association of the amine complexes at an average concentration at which the dielectric constant measurements were made is approximately 2.8. Comparison of the values of a for the amine complexes with those for other complexes shows that the solutions contain much less polar species.

<u>Reactions carried out to investigate the relative donor strengths</u> of various ligands to phenylethynylgold

About 0.5g. complex was dissolved in benzene or 1,2dimethoxyethane and a slight excess over one equivalent of the appropriate ligand was added. After removal of volatile material (in vacuo), the complex remaining was recrystallised and identified by its melting point and infra-red spectra.

95.

				,		
(1)	RNH2 octy1 ⁿ	+	PEt3	\rightarrow	RPEt ₃ [*]	+ octyl ⁿ NH ₂
(2)	RAsEt 3	+	PEt3	\rightarrow	RPEt ₃ *	+ AsEt ₃
(3)	RCN <u>p</u> -tolyl	+	PEt3	\rightarrow	RPEt [*]	+ p-tolyl NCt
(4)	RP(OPh)3	+	PEt ₃	\rightarrow	$\operatorname{RPEt}_{3}^{*}$	+ P(OPh) ₃
(5)	$RNH_2 \text{ octyl}^n$	+	AsEt ₃	\rightarrow	$RAsEt_3$ *	+ octyl ⁿ NH ₂
(6)	RNH ₂ pentyl ⁿ	¹ +	p-tolyl NC	\rightarrow	RCNp-toly1*	+ pentyl ⁿ NH ₂
(7)	RAsEt 3	÷	p-tolyl NC	\rightarrow	RCNp-toly1*	+ AsEt ₃ †
(8)	$_{\text{RNH}_2} \text{ octyl}^n$	+	SbEt ₃	\rightarrow	RSbEt ₃ +	+ octyl ⁿ NH ₂
(9)	RAsEt 3	+	SbEt ₃	\rightarrow	RAsEt ₃ uncl	nanged
(10)	RCN <u>p</u> -tolyl	+	P(OPh) ₃	\rightarrow	$\mathbb{RP}(OP_h)_3^{\mathbf{X}}$.	+ p-tolylNC +
(11)	RPEt ₃	. +	PPh3	\rightarrow	RPPh ₃ ‡ -	+ PEt ₃ +
(12)	RPEt 3	+	P(OPh)3	\rightarrow	RPEt ₃ uncl	nanged
(13)	$RNH_2 \text{ octyl}^n$	+	Pr ⁱ NH ₂	\rightarrow	RNH ₂ Pr ⁱ *	+ octyl ⁿ NH ₂
(14)	$RNH_2 \text{ octyl}^n$	+	NH ₃	\rightarrow	RNH ₃ ≠↓	+ octyl ⁿ NH ₂
(15)	${\tt RNH}_2 \text{ octyl}^n$	+	pyridine	\rightarrow	\mathbb{RNH}_2 octyl ⁿ	unchanged
		R	= phenylethyr	nylgold,	PhC:CAu	
		X	identified by	m.p. and	l I.R. spectr	um
	an an tha an an an an An Anna an Anna Anna Anna An Anna Anna	‡	identified by	I.R. spe	ectrum	
		†	identified by	smell		

identified by smell

TERTIARYBUTYLETHYNYLGOLD AND SOME DERIVATIVES

Preparation of t-butylethynylgold ButC:CAu

This was prepared by the method used for phenylethynylgold. t-Butylacetylene (2.06g., 2.7cc., 0.0258 mole) was added to the solution after the reduction of gold(III) chloride (lOg. hydrate), followed by sodium acetate (25g.) in water (100cc.). A pale yellow precipitate formed and gradually thickened. Water was then added until the flask was almost full, and after stirring for 30 minutes the solid was filtered off, washed with water and dried in vacuo. Yield: 6g., 77%). t-Butylethynylgold can be crystallised from propanol as small fine crystals, decomp. above 150°, or from hexane (by cooling a solution over acetone/ (Found: C, 26.2; H, 3.3; Au, 70.6, 70.5; M, ebullio-CO_). scopically in 1.48, 2.46, 2.93, 3.47, 4.47 wt. % benzene, 1166, 1192, 1154, 1118, 1112. (C₆H₆Au)₄ requires C, 25.9; H, 3.3; Au, 70.9%, M, 1113), It is soluble in benzene, chloroform and hexane, slightly soluble in acetone and insoluble in methanol.

Frequently on recrystallising the crude material, some remained insoluble in the boiling solvent and was recovered as a purplish-white powder, insoluble in most organic solvents. It appears to be a more polymeric form of t-butylethynylgold, and on addition of trimethylphosphine to an ether suspension it afforded a complex identical (mixed m.p. and infra-red spectrum) with that from the tetramer.

t-Butylethynyl(ammine)gold Bu^tC:CAuNH₃

Addition of excess ammonia to $(Bu^{t}C:CAu)_{4}$ (2g.) suspended in acetone (20cc.) gave a white precipitate which dissolved on warming. The hot filtered solution deposited colourless feathery needles (1.0g.), and a further 0.5g. were recovered after concentration of the mother liquor, decomp. above 155°. (Found: C, 25.4; H, 4.2; Au, 66.5, 66.6; N, 4.66. $C_{6}H_{12}AuN$ requires C, 24.4; H, 4.1; Au, 66.8; N, 4.74%. A solution in boiling benzene loses ammonia to give t-butylethynylgold. <u>t-Butylethyn (trimethylphosphine)gold</u> $Bu^{t}C:CAuPMe_{3}$

Addition of trimethylphosphine (3.65 m mole in ether 5cc.) to a solution $(Bu^{t}C:CAu)_{4}$ (lg., 0.9 m mole) in benzene (lOcc.) discharged the yellow colour immediately. Removal of benzene and recrystallisation from a 1,2-dimethoxyethanehexane mixture afforded colourless plates, lg., m.p. 194-196° (decomp). (Found: C, 30.4; H, 5.1; Au, 55.7; 55.8. $C_{9}H_{18}AuP$ requires C, 30.5; H, 5.1; Au, 55.6%).

t-Butylethynyl(triphenylphosphine)gold Bu^tC:CAuPPh₃

Addition of triphenylphosphine in benzene to a benzene solution of t-butylethynyl(ammine)gold, followed by removal of benzene, and crystallisation of the solid remaining from a 1,2dimethoxyethane-hexane mixture afforded Bu^tC:CAuPPh₃ as colourless crystals, m.p. 192-195[°] (decomp.). (Found: C, 52.3; H, 4.4. C₂₄H₂₄AuP requires C, 51.4; H, 4.3%).

Other reactions of phenylethynylgold

The reaction between phenylethynylgold and phosphine

Phenylethynylgold (2.166g.) was suspended in benzene (25cc.) and phosphine (from Zn_3P_2 and dil. H_2SO_4) bubbled through the stirred suspension, which turned orange, though no colour was imparted to the benzene. The suspension was filtered (N_2 atmosphere) and the solid washed with benzene, and analysed: C, 2.9; H, 2.1; Au, 84.8, 85.4%. H_2AuP requires C, 0.0; H, 0.88%; Au, 85.7%. Yield: 1.59g., 95.2%. It was insoluble in all common solvents and was fairly air-stable. All the absorption frequencies in the infra-red spectrum could be attributed to phosphorus-hydrogen linkages.

The filtrate from the reaction was distilled until half the benzene and any free phosphine had been removed. It was then added to ammoniacal copper(I) chloride solution giving phenylethynylcopper (1.09g.), corresponding to 95% of the phenylethynyl group present in the starting material. The benzene layer, now free from phenylacetylene was reduced to dryness, there being no residue. This experiment is consistent with the reaction

PhC:CAu + $PH_3 \rightarrow PhC:CH + (AuPH_2)_x$ t-Butylethynylgold and phosphine react similarly.

The reaction between quaternary ammonium or phosphonium iodides and PhC:CAu

Phenylethynylgold and tetraethylammonium iodide

Phenylethynylgold (4g., 13.4 m_mole) was suspended in ethanol (10cc.). On addition of tetraethylammonium iodide (3.5g., 13.6 m_mole) the suspension was replaced by a mixture of a grey crystalline precipitate and gold powder, which was then filtered off and extracted with hot acetone, gold (1.3g., 50%), and colourless needles (1.8g.) which crystallised from the acetone on cooling, being recovered. The colourless needles contained: C, 58.9; H, 5.64; Au, 32.7, 32.9%, but no iodine, and had m.p. 198-200° (decomp.). The ethanolic filtrate from the reaction afforded unreacted tetraethylammonium iodide (1.3g., m.p. 312°).

A similar reaction gave fine colourless crystals (m.p. 174-176° (decomp.) from acetone) containing, on analysis, Au, 26.1, 26.8%. The compound from the two first had absorption due to acetylenic stretching, at approx. 2141 and 2096 cm.⁻¹. The second compound had one only, at 2146 cm.⁻¹.

The compound (PhC:C)₂AuNEt₄, if formed would require C, 54.4; H, 5.7; Au, 37.2%.

Phenylethynylgold and trimethylphenylphosphonium iodide

On addition of trimethylphenylphosphonium iodide (2.8g., 10 m_mole) in methanol (20cc.) to a suspension of phenylethynylgold (3g., 10m mole) in methanol (10cc.), the phenylethynylgold dissolved and gold (1g.) was precipitated and filtered off. Removal of the methanol from the filtrate, and recrystallisation of the solid thus obtained from an acetone-ethanol mixture afforded small white needles, 1g., m.p. 149-150°. (Found: C, 59.7; H, 4.4; Au, 25.5. (PhC:C)₂AuPMe₃Ph, if formed, requires, C, 54.3 H, 4.2; Au, 35.8%.). This solid had one acetylenic frequency at 2144 cm.⁻¹.

DISCUSSION

CHAPTER VI

DISCUSSION

Copper(I), Silver(I) and Gold(I) derivatives of monoacetylenes RC:CM.

The structure of the polymeric copper and silver ethynyls RC:CCu(Ag) is still unknown. With the exception of t-butylethynylcopper which is octameric ebullioscopically in benzene, this figure agreeing with the cryoscopic values obtained by other workers.^{92,141} all the copper acetylides are insoluble in all common organic solvents other than those with which they react (.e.g. some amines). A possible structure has been suggested.¹ as described earlier, but any experimental evidence may not be forthcoming for some time since it is exceedingly difficult to obtain sufficiently well-defined crystals suitable for X-ray crystal structure determinations. The copper and silver ethynyls were prepared by the wellestablished methods, as powders which required prolonged pumping, in vacuo, to dry them. The arylethynylcopper compounds were in general slightly soluble in isopropylamine and diethylamine; the complexes formed are only weak and extraction of phenylethynyl- and mesitylethynyl copper by these amines in a Soxhlet apparatus in a nitrogen atmosphere, afforded

micro-crystalline material, extraction being slower with the secondary amine. Further extractions with isopropylamine by Mr. P. W. R. Corfield¹⁴² of this department afforded similar crystals which unfortunately were "twinned" and, thus, unsuitable for detailed X-ray study. Repetition of the work of F. Straus who obtained¹³³ phenylethynylcopper as "a fine mass of yellow crystalline needles" from copper(II) phenylpropiolate in the presence of various amines only provided powdery solids. Some of the alkylethynylsilvers are moderately soluble in

certain organic solvents, in particular in chloroform, carbon tetrachloride and acetone. t-Butylethynylsilver, the most soluble silver ethynyl, was extracted with benzene, cyclohexane, isopropylamine and diethylamine, some decomposition resulting in the higher boiling solvents, but in all cases feathery needles were obtained.

The only goldethynyl hitherto described is the unstable acetylide Au_2C_2 . A general method developed to prepare goldethynyls $(R \cdot C:CAu)_x$ consists of adding the acetylene and sodium acetate solution, to a freshly reduced solution of gold(III) chloride containing sodium bromide, and has been used to prepare the new compounds, phenylethynylgold(I) and t-butylethynylgold(I). Phenylethynylgold, a yellow powder, resembles its copper analogue in appearance and is presumably a similar co-ordination polymer. It is insoluble in common organic solvents, but differs

in being soluble in and forming isolatable complexes with a considerably greater variety of donor solvents, and is no doubt a typical gold(I)ethynyl. t-Butylethynylgold is also similar to its copper analogue in that it is soluble in many organic solvents, but whereas t-butylethynylcopper is octameric ebullioscopically in benzene the pale-yellow gold compound is tetrameric. If the co-ordination number of the gold is two, as is most common for gold(I), the compound is likely to have the square structure

An insoluble form of t-butylethynylgold was also obtained as a pale purplish-white powder (the colour almost certainly being due to a trace of metallic gold) which is probably a polymer of similar structure to that of phenylethynylgold:



The infra-red spectra of the two forms were almost identical, with slight difference in the acetylenic stretching region (Table 8, page 128).

These organic group IB ethynyls could not be made to explode, including the pale yellow <u>paranitrophenylethynylsilver</u> which is reported¹³⁴ to explode on heating. It appears that only the acetylides, M_2C_2 , (M = Cu, Ag, Au) and the derivatives of monohaloacetylenes XC:CM (X = halogen, M = Cu, Ag, no gold compounds are yet described) possess this property.

Co-ordination complexes of copper(I), silver(I) and gold(I) ethynyls

Ethynylcopper(I) complexes with tertiary phosphines

Difficulties due to the inconveniently high solubilities of the triethylphosphine complexes reported previously⁹² were largely overcome by the use of trimethylphosphine, the more symmetrical molecule usually resulting in higher melting points and lower solubilities.

105

The yellow monophosphine complexes (Table 2a) prepared

TABLE 2a

Degree of association of mono-phosphine complexes

	cryoscopically			ebullioscopical		
	in benz wt.%	ene <u>n</u>	in nitro wt.%	obenzene <u>n</u>	wt.%	<u>n</u>
CH ₂ :CMeC:CCuPMe ₃	1.77 2.18 2.49 2.66	3.34 3.28 3.37 3.39	1.23 1.90 2.34 2.64 2.85	2.12 2.31 2.24 2.16 2.04		
					in ace	tone
Bu ^t C:CCuPMe3	2.78 3.35 3.84 4.20	2.67 2.79 2.73 2.74	1.65 1.94 2.29 2.49	2.01 2.04 1.99 2.05	0.50 1.00	2.4 2.6
					in ben	zene
PhC:CCuPMe3	0.91 1.81 2.41 2.91 3.07 6.36	2.60 2.75 2.80 2.79 2.70 2.70	0.39 0.41 0.53 0.61 0.73 0.99	1.94 1.99 2.04 2.06 1.88 2.05	1.59 1.80 3.01 3.80	3.18 2.62 2.67 2.62
PhC:CCuPEt ₃	0.68 1.16 1.27 2.00 2.63	2.93 2.70 2.54 2.64 2.70	0.44 0.97 1.50 2.05	2.00 2.02 1.89 2.03		
PhC:CCuPPh ₃	too sr	aring	ly solubl	le		and an ann an Airtean Airtean an Airtean Airtean an Airtean
pN0 ₂ C ₆ H ₄ C:CCuPPh ₃	1.14 1.77 2.18 2.44	2.10 2.07 2.02 1.96	0.39 0.49 0.64 0.76	1.12 0.94 0.92 1.14		

by addition of the required amount of the tertiary phosphine to a suspension of the insoluble copper ethynyl in an organic solvent, followed by filtration and crystallisation, are usually dimeric in nitrobenzene and slightly associated in benzene (between 2.6 and 3.3). The nitrobenzene solutions may well contain dimer molecules of structure

 $R C \equiv C - Cu PR'_{3}$ $\int T$ $Cu - C \equiv CR$ $R'_{3}P$

in which case the copper atoms have the co-ordination number three. It could also be possible that the solvent acts as a donor as well as ethynyl groups thus allowing the more usual four-co-ordination. <u>paraNitrophenylethynyl(triphenylphosphine)</u> is exceptional in being monomeric in nitrobenzene and dimeric in benzene. It is unlikely that nitrobenzene could behave as a donor of sufficient strength to break the dimer. This complex was obtained in two modifications. In non-polar solvents it formed orange solutions, from which orange crystals were recovered; on allowing these crystals to stand in contact with ethanol in which they are insoluble, they gave a powdery red material. Unfortunately yellow-green phenylethynyl(triphenylphosphine)copper is too sparingly soluble for molecular weight



determinations. These triphenylphosphine complexes are airstable unlike the trialkylphosphine complexes. Propynylcopper did mot form a triphenylphosphine and did not readily yield isolable phosphine complexes. The complexes isolated were very air sensitive and exceedingly soluble in hexane, but they dissociate in the absence of an excess of the phosphine. The stability of the trimethylphosphine complexes (Me₃PCuCiCR,R = Me, Bu^t, Ph,isopropenyl) isolated increases in the order Me \ll isopropenyl \leq Bu^t \leq Ph.

The structure (Figure 5 opposite) of the complex PhC:CCuPMe3 has been examined crystallographically by Dr. H.M.M. Shearer and Mr. P.W.R. Corfield of this department. It contains tetrameric units (PhC:CCuPMe3) each with a zig-zag chain of four copper atoms; two phosphine molecules are bound to each of the terminal copper atoms, the central two atoms being/coordinated to phosphorus. Two of the four acetylene groups have a different environment from the other two. Apart from the phosphine groups (the methyls not being shown in the figure) which lie above and below the plane of the paper the structure is nearly flat. The acetylene groups are bent, particularly those (C_9 and C_{10}) which form "broadside-on" π bonds to copper. The other kind are clearly σ -bonded to the copper together with some "end-on" π bonding. The environment of Cu₂ is roughly (with respect to the phesphines and Cu₁ end² tetrahedral) trigonal with respect to the phosphines and carbon atoms C₁ and The copper atoms Cu_1 appear to have five ligands in a plane Cq.

but if the possibility of Cu-Cu bonding is ignored although the Cu-Cu distances are close to those in metallic copper, then Cu₁ can be regarded as trigonally co-ordinated and Cu₂ as tetrahedrally co-ordinated.

Several bisphosphine complexes were isolated though they were more difficult to purify. They support further the view that ethynylcopper(I) complexes are co-ordination polymers in which polymerisation is reduced by tertiary phosphines successfully competing with ethynyl groups as ligands. When two phosphine groups are co-ordinated to each copper atom the complexes are monomeric and 3 covalent. (Table 2b).

Degree of associ	ation cry	oscopically,	bisphospl	hine
	Complex	es		
	in benz	ene	in nitrob	enzene
	wt.%	<u>n</u> - 11 - 11 - 11 - 11 - 11 - 11 - 11 -	wt.%	n
Bu ^t C:CCu(PMe ₃) ₂	0.31 0.71 1.41 1.85	1.00 0.98 0.98 0.98	0.18 0.62 1.13 1.47	0.43 0.75 0.88 0.84
PhC:CCu(PMe ₃) ₂	0.63 0.86 1.17 1.71	0.98 1.12 1.35 1.37	0.53 0.78 1.09 1.34	0.93 1.00 0.95 0.98
\underline{p} -NO ₂ C ₆ H ₄ C ^{!} CCu(PPhEt ₂) ₂	0.61 0.94 1.28	0.57 0.63 0.69	0.32 0.50	0.38 0.26

TABLE 2b

The t-butylethynyl complex dissociates in nitrobenzene solution, though if nitrobenzene was acting as a donor and

partially replacing the phosphine, the same cryoscopic result would be obtained. However the <u>paranitrophenylethynyl</u> complex is dissociated in both benzene and nitrobenzene. Dissociation could also be responsible for the t-butyl and phenyl complexes melting over a range of several degrees.

Progressive addition of tertiary phosphine causes a marked lightening of colour. For example, t-butylethynylcopper which is orange-red yields a yellow-green mono(trimethylphosphine) complex and a colourless bisphosphine complex.

In the case of the aliphaticethynylcoppers and of phenylethynylcopper, to dissolve a certain amount of the ethynyl in benzene, one equivalent of an aliphatic phosphine is required. paraNitrophenylethynylcopper appears to be exceptional and two equivalents of the phosphine are required - apparently the bisphosphine complex is more stable than the monophosphine complex (which was never isolated) and is formed in preference. paraBromophenylethynylcopper, although not studied in detail, behaves similarly, two equivalents of triethylphosphine being required to dissolve a given amount of the ethynyl. It seems that the electrophilic nature of the nitro and bromo groups draws electrons from the copper atom which can then more readily accept two phosphine molecules. If this is a likely explanation then mesitylethynylcopper might be expected to form a 1:1 complex as phenylethynylcopper does, and in fact a known weight of mesitylethynylcopper dissolved readily in benzene on addition of one equivalent of trimethylphosphine.

One trisphosphine complex PhC:CCu(PMe₃)₃ was isolated, but it dissociated in solution and had a strong smell of trimethylphosphine. It is presumably monomeric and analagous to the previously reported³⁴ isocyanide complex PhC:CCu(CNp-tolyl)₃.

The presence of ethynyl groups in two different environments in the complex $(PhC:CCuPMe_3)_4$ suggests that it should be possible to prepare a complex of the type $[R \ C \ CCu, \ R^{\circ}C:CCu(PMe_3)_2]_2$ in which two ethynyl radicals are incorporated into the molecule. This was achieved in one case $g(=Gu^c)$, where $R = Ph_{,/}$ and the complex had a sharp melting point between those of the two complexes $PhC:CCuPMe_3$ and $Bu^tC:CCuPMe_3$. An equimolar mixture of these two complexes melted thirty degrees lower with decomposition and so confirmed that a new compound and not the equimolar mixture had been prepared.

TABLE 2c

Association, cryoscopically, of PhC:CCu Bu^tC:CCu 2PMe3

in benzene		i	in nitrobenzene		
wt.%	<u>n</u>		wt.%	n	
1.69	1.44	1	14	0.96	
2.54	1.44	1	71	1.00	
3.5	1.42	2	.05	1.02	
3.39	1.38		.28	0.96	

Molecular weight measurements (Table 2c) correspond to those found for [PhC!CCuPMe3]4 and presumably the complex has a similar

structure to the tetramer in which two of the phenyls are replaced by t-butyl groups.

It is interesting to note that when 2 mols. of triethylphosphine are added to an equimolar suspension of propynylcopper and phenylethynylcopper in benzene, the propynylcopper remained unchanged, phenyl_ethynylbistriethylphosphinecopper being formed. To prepare a mixed complex of the above type it therefore seems necessary that there must be a minimum difference between the stabilities of the related constituent complexes.

The diacetylene complexes $PhC:C.C:CCuPR_3$ (R = Me, Et, Pr^n) prepared were unexpectedly found to be less air-sensitive than the monoacetylene compounds. In contrast the diacetylenes (e.g. PhC:C.C:CH which decomposes above $0^{\circ}C$) are much less stable than the monoacetylenes. Whereas $PhC:C.C:CuPEt_3$ reverted to yellow amorphous PhC:CCu after exposure to air for a few days the corresponding butadiynyl complex decomposed slowly over several months and the trimethylphosphine complex appeared unchanged after a year. Molecular weight measurements (Table 2d) indicate that they are associated similarly to the monophosphine complexes.

An interesting complex with a Cu:P ratio 2:3 was characterised. It was isolated in an unsuccessfull attempt to prepare a bisphosphine complex over which it would therefore seem to possess extra stability. It loses triethylphosphine

TABLE 2d

Association, cryosc	opically, of	phenylbutad	iynyl comple	xes
	PhC:C.C:CCul			
	benzo	ene	nitroben	zene
	wt. %	<u>n</u>	wt.%	<u>n</u>
PMe ₃	t	oo sparingly	soluble	
PEt ₃	1.10	3.27	1.13	1.97
	1.40	3.29	1.75	2.04
	1.71	3.05	2.15	2.11
	1.77	3.23	2.43	2.13
PPr ₃ ⁿ	2.52	3.36	1.05	2.00
	3.78	3.19	1.40	2.00
	4.54	3.11	1.68	2.04
	5.09	3.01	2.11	1.95
(PEt ₃) <u>3</u> 2	0.91	0.71	0.57	0.91
	1.31	0.51	0.99	1.03
	1.64	0.63	1.49	0.95
	1.82	0.69	1.87	0.96

on standing in air or on heating a solution in propanol, giving the monophosphine complex and from an x-ray study Dr. Shearer and Mr. Corfield confirm this Cu:P ratio. They report that the space group is similar to that of $(PhC:CCuPMe_3)_4$ and the data are consistent with a structure different from that of $PhC:C \cdot CuPMe_3$ only in that three instead of two phosphine groups are bound to the termihal copper atoms $(Cu_2 \text{ in figure 5})$ of the Cu_4 chain. In nitrobenzene this complex dissociates as a dimer [i.e. $(PhC:C:CCu)_2(PEt_3)_3$, with two copper atoms per molecule] and it is more extensively

dissociated in benzene. (Table 3d).

Silverethynyl complexes

Little work was carried out on silver complexes as previous preliminary investigations¹ showed them to be similar to their analagous copper complexes. They also seem to be less easy to purify than the copper compounds which is probably chiefly due to the fact that they are quite sensitive to light. Phenylethynyl(trimethylphosphine)silver PhC:CAgPMe₃ which is less soluble than its copper analogue, is dimeric ebullioscopically in benzene.

Co-ordination compounds of gold(I) ethynyls

The first gold(I)ethynyl complex, the phosphine complex PhC:CAuPEt₃, was prepared¹⁴⁶ from bromo(triethylphosphine)gold(I) and phenylethynyllithium. The reaction between phenylethynylgold and triethylphosphine gave an identical complex confirming the isolation of phenylethynylgold. By similar direct reactions between phenylethynylgold and donor molecules it has been possible to prepare a range of complexes many of which could not have been obtained by the reaction of an organolithium reagent on a gold(I) halide. In this way complexes of phenylethynylgold with ammonia, primary and secondary amines, tertiary phosphines arsines and stibines, phosphites and <u>iso</u>cyanides were prepared.

The complexes with triphenylphosphine, triphenylphosphite, triethylarsine, n-butylisocyanide and o-ethylphenylisocyanide,

are monomeric in benzene as would be expected for gold(I) compounds. The complex with triethylstibine was air-sensitive and unstable, and no molecular weight measurements were carried out, but it would be surprising if it were not monomeric too. Phenylethynyl(triethylphosphine)gold has been previously reported to have an association of 1:1 to 1.2 in nitrobenzene and about 1.3 in benzene¹²² A diphosphine complex was also obtained, in which 2 molecules of phenylethynylgold give PhC:CAu \leftarrow PPh₂CH₂CH₂Ph₂P \rightarrow AuC:CPh This complex was less soluble than the other phosphine complexes, being insoluble in cold benzene.

These monomeric complexes, in which the Au-C and P-Au bonds are co-linear, are particularly suited to dipole moment measurements. The dipole moments (Table 3) are with the exception of the triphenylphosphite complex, in the range 6.2 - 7.1 D.

TABLE

Dipole moments of some phenylethynylgold(I) complexes

Ph ₃ PAuC: CPh	6.7
(PhO) ₃ PAuC:CPh	4.4
Et 3AsAuC: CPh	6.2
Bu ⁿ NCAuC:CPh	6.9
<u>o</u> -c ₂ H ₅ C ₆ H ₄ NCAuC : CPh	7.05
* Et 3PAuC: CPh	6.55

included for comparison

X	Et ₃ PAuMe	5.5
¥	Ph ₃ PAuMe	5.6
X	Et 3PAuEt	5.5

* measured by Dr. P. Dixon

The triphenylphosphite complex is peculiar, particularly if the moment of triphenylphosphite itself is in the same direction as triphenylphosphine, i.e. P - C which is likely. The effect of changing Et_3P by Ph_3P causes an increase in moment of 0.15D which is similar to the increase on going from Et_3PAuMe to Ph_3PAuMe , but which is not in the expected sense, as triphenylphosphine is known to be a better electron-acceptor than triethylphosphine and this would tend to decrease the dipole. The <u>o</u>-ethylphenylisocyanide complex might also be expected to have a lower moment than n-butylisocyanide complex instead of the observed increase of 0.15D. In this region of 7D the observed moments are particularly accurate, having an error not greater than 0.02D.

Phenylethynylgold forms a stable colourless ammine $(PhC:CAuNH_3)_x$ in the presence of ammonia analagous to the less stable phenylethynyl(ammine)copper. It is insoluble in all common organic solvents. Crystalline complexes with four primary aliphatic amines were isolated. The isopropyl and n-pentylamine complexes were too sparingly soluble for molecular

weight determinations suggesting they may be associated in some way and the n-octyl- and n-nonylamine complexes had apparant degrees of association, which increased with concentration, in the range 2.6 - 4.1. Ebullioscopic methods are impracticable since the amine complexes, the least thermally stable of the phenylethynylgold complexes isolated, decompose fairly quickly in solution on warming and the solutions darken slowly at room temperature. Since benzene solutions of these complexes evidently contain two or more molecular species it is not possible to calculate dipole moments, however dielectric measurements of their benzene solutions indicate the presence of species which are much less polar than the other complexes studied. The acetylenic stretching frequencies (see page 117) in these complexes are similar to those found in the monomeric complexes, and suggest that the acetylenic group is not involved to any appreciable extent in co-ordination to gold to make up a polymeric structure. The co-ordination number of the gold is evidently greater than two.

Complexes with the secondary amines, di-n-butylamine and piperidine were characterized. The piperidine complex was soluble in benzene, and solutions of the di-n-butylamine complex turned red very readily, depositing a black sludge and rendering them useless for molecular weight or dipole moment determinations. The only tertiary amine to yield an isolatable complex was pyridine, and the complex reverted to phenylethynylgold when placed in organic solvents containing no excess pyridine. Aliphatic tertiary amines dissolve benzene suspensions of phenylethynylgold giving solutions which darken readily and deposit gold.

The relative donor strength of the various ligands, mentioned so far, towards phenylethynylgold was established by a series of displacement reactions given in detail on page **96**. The resulting order was found to be

PPh₃ > PEt₃ > P(OPh)₃ > ptolylNC > AsEt₃ > SbEt₃

> ammonia, > tertiary amines

It is also almost certain that $NH_3 > RNH_2 > R_1NH$, the secondary amine complexes showing a markedly lower stability than primary amine complexes. It is interesting that triphenylphosphine will replace triethylphosphine from its complex, triphenylphosphine being a stronger π acceptor; the series in fact appears to be largely determined by the extent to which the ligand is able to accept electrons from the gold <u>d</u> orbitals. In this respect, it is interesting to note the position of the isocyanide ligand and that triethylstibine will replace amines.

Several other ligands were tried as donors to phenylethynylgold. Trimethylphosphite and triethylphosphite gave complexex which were oils at room temperature. Diethylsulphide, diphenylacetylene methyl and phenylcyanide and carbon monoxide at atmospheric pressure did not yield complexes. Phenylethynylgold was extracted slowly by dimethylsulphide (using a Soxhlet apparatus) but the resulting solution continually deposited gold. Phosphorus(III) chloride, which has been seen to be a better co-ordinating ligand than the alkylphosphines to some metals in low valency states¹⁴³ due to its greater acceptor properties, did not appear to form a complex with phenylethynylgold.

Since fairly weak donors such as triethylstibine form complexes with phenylethynylgold, it might be expected that, quaternary ammonium and phosphonium iodides would give salts of the type $[Et_4N]^+$ [PhC:CAuI]⁻ in which the iodine acts as a donor. Phenylethynylgold does in fact dissolve in cold alcoholic solutions of these iodides, but the white crystalline solids obtained from these solutions (together with some gold metal) did not contain iodine. Analytical data did not agree with another possible compound $[R_4N(P)]^+$ [(PhC:C)₂Au]⁻ and more than one product may result from this reaction.

Phosphine, itself, did not give a complex of phenylethynylgold but acted as an acid to give polymeric $(AuPH_2)_x$ which is insoluble in all common inert solvents, and darkens on keeping. In view of this and of recent work¹⁴⁴ on the reaction of palladium complexes with secondary phosphines, it would be interesting to look at the reaction of phenylethynylgold with primary and secondary phosphines.

A few complexes of the other gold ethynyl prepared,

t-butylethynylgold, were isolated. Similar complexes to those of phenylethynylgold were obtained with ammonia, trimethyland triphenyl- phosphines. The ammine, Bu^tC:CAuNH₃ is moderately soluble in various organic solvents, in contrast to the phenyl analogue, but unfortunately a solution in benzene loses ammonia on boiling, and its degree of association could not be determined.

CHAPTER VII

INTERPRETATION AND DISCUSSION OF INFRA-RED SPECTRA

The acetylenic stretching frequencies

Disubstituted acetylenes absorb light in the range¹⁴⁵ 2190 - 2260 cm⁻¹. Co-ordination between acetylenic groups and the metal atomsshould result in a reduction of the stretching frequency because of electron flow from the occupied π u bonding orbitals of the acetylene to the metal and because of electron flow from occupied metal <u>d</u> orbitals to unoccupied π g acetylene antibonding orbitals. Both these effects should reduce the acetylenic bond order, and hence the stretching frequency. Effects of this type have been described previously; for example, the acetylenic stretching frequency of ditertiarybutylacetylene is reduced by about 200 cm⁻¹ when it is π -bonded to platinum,¹⁴⁶ and similar effects have been found in heavy metal complexes of olefins.¹⁴⁷

In agreement with the view¹ that the alkyl and arylethynylcopper(I) compounds, $(RC:CCu)_x$ are co-ordination polymers in which there is substantial back co-ordination from filled copper 3<u>d</u> orbitals to the acetylenic antibonding orbitals, these compounds have been found to absorb in the region 1920-1960 cm.⁻¹ which is 270 - 300 cm.⁻¹ lower than the range characteristic of disubstituted acetylenes. (Table 4).
TABLE 4

Stretching fre	quencies of t	he acetylenic group in	RC:CM
R۰	M = 1 Hg	Cu	Ag
Me•	2156	1955	2061
Et.	2165	1952 ca.	2040 ^{a,c}
Pr ⁿ .	2164 ^a 2175 ^b	1942	2045
Bu ^t .	2180 2146 2188 2154	absent	2055
Ph.	2149	1933	2055
pNO2C6H4.	$absent^{a,c}$	1929	2042
mesityl-	2142	1933 ^c (<u>v.w</u> .)	2026
pBrC6H4.	2160	1942	2048
PhC:C.	2204 2098	2180 ^{a,c}	2190

measured in ^a KBr disc, ^b benzene solution, ^c Nujol mult

It can be seen that this effect is less marked in the corresponding silver compounds whose acetylenic frequencies are in the range 2020-2060 cm.⁻¹, a reduction of only 150-200cm.⁻¹. The acetylenic frequencies of the corresponding ethynylmercury(II) compounds are included in Table 4 for comparison as examples of disubstituted acetylenes containing 'normal' ethynyl-metal bonds. On substituting copper and silver for mercury the decrease in $\Im(C:C)$ is about 200 and 100 cm.⁻¹ respectively. In the case of t-butylethynylcopper(I) no absorption that could be due to an acetylenic vibration could be detected either in a potassium bromide disc or in solution in tetrachlorethylene. This could be connected with the mass of a copper atom being fairly near that of the butyl group. However, no similar frequency could be detected in the case of bisp-nitrophenylethynylmercury and in the case of phenylbutadiynyl-copper and -silver the frequencies observed are obviously due to one of the two acetylene groups which is apparantly not involved in the structure of the co-ordination polymer.

Phenylethynylgold absorbs very weakly at 1973 cm.⁻¹ which is about 175 cm.⁻¹ lower than the absorption in the mercury compound and near that in phenylethynylcopper. t-Butylethynylgold has three weak absorptions between 2030 and 1950 cm⁻¹ (see Table 8, page 128) which are similarly lower. Co-ordination between the metal and the acetylene decreases in the sense Cu > Au > Ag in these uncomplexed ethynyls, (RC:CM)_x.

In the case of the ethynylcopper(I) compounds complex formation with a tertiary phosphine raises \Im (C:C) by nearly 100 cm⁻¹ from that observed in (RC:CCu)_x, and it is peculiar that the frequency of the acetylenic absorptions is much the same whether the complex contains one, two or more molecules of phosphine. Table 5 lists the acetylenic frequencies of the copperethynyl complexes for which two or more \Im (C:C) bands were observed in the spectra of the solid complexes pressed in a potassium bromide disc.

TABLE 5

Acetylenic stretching frequencies of	ethynylcop	per(I) comp	lexes
Bu ^t C:CCuPMe ₃	2050 (m^{-1} (<u>vw</u>)	
PhC:CCuPMe ₃	2019	(<u>w</u>), 2045	(<u>m</u>)
PhC:CCu(PMe ₃) ₂	2035	(<u>m</u>), 2048	(<u>m</u>)
$PhC:CCu(PMe_3)_3$	2034	(<u>s</u>), 2051	(<u>vw</u>)
PhC:CCuPEt ₃	a 2018 2020	$\left(\frac{s}{s}\right)$, 2048	(<u>m</u>)
PhC:CCuPPh ₃	2043	(<u>m</u>)	
p-NO ₂ C ₆ H ₄ C:CCuPPh ₃	2015	(<u>'s</u>)	
p-NO ₂ C ₆ H ₄ C:CCu(PPhEt ₂) ₂	2025	(<u>s</u>)	
PhC:C·C:CCuPMe3	2161	(<u>s</u>), 1981	(<u>m</u>)
PhC:C.C:CCuPEt ₃	ъ 2159 216 7	$(\underline{s}), 1988$ $(\underline{s}), 1988$	$\binom{\underline{m}}{\underline{m}}$
$(PhC:C\cdotC:CCu)(PEt_3)_3$	a 2165 2171	$(\underline{s}), 2008 \\ (\underline{s}), 2009$	$\left(\underline{m} \right)$ $\left(\underline{m} \right)$
$PhC:C \cdot C:CCuPPr_3^n$	ъ 2164 2166	$(\underline{s}), 1984$ $(\underline{s}), 1986$	$\binom{\underline{m}}{\underline{m}}$
Bu ^t C:CCu(PhC:CCu)(PMe ₃) ₂	2062	(<u>m</u>)	
* Spectra measured in KBr discs exc	ept: ^a in b	enzene, ^b i	n
tetrachlorethylene.			
cf. PhC:CAgPMe3	2075	(<u>w</u>)	
PhC:CAuPMe3	2095	(<u>m</u>)	
<u>vw</u> very weak, <u>w</u> , weak, <u>m</u> , medium,	s, strong al	bsorption	
The silver and gold complexes PhC:C	Ag(Au)PMe 3	are added s	since
they show that $\mathfrak{J}(C:C)$ increases in	the series]	PhC:CMPR3 i	n the
sense $M = Cu \leq Ag \leq Au$ and that the	change from	n (RC:CC11).	- to

RC:CCuPR₃ (about \bigwedge^{100} cm⁻¹) is much greater than that from (RC:CAg)_x to R C:CAgPR₃ (about 20 cm⁻¹) but not as great as that from (RC:CAu)_x to monomeric RC:CAuPR₃ (about 130 cm⁻¹).

In the butadiynyl complexes it can be seen that one of the two acetylene groups is not involved in co-ordination, and the other is involved slightly more than the acetylenic group in the mono-ethynyl complexes.

Two series of copper complexes $RC:CCu(PMe_3)_n$ (R = Ph and $CH_2:CMe\cdot$) were studied in some detail. Trimethylphosphine was added to the 1:1 complex dissolved in benzene, so that the ratio of phosphine to RC:CCu was increased to 2:1, 3:1 and 4:1 (i.e. n = 2,3 and 4). The acetylenic absorptions observed for these different solutions containing increasing P:Cu ratios are shown in Table 6. Extinction coefficients are included in parenthesis.

Effect of P:(Cu ratio on J(C	C) in benzene	solution of	RC:CCu(PMe3)n_
R•	n = 1	n = 2	n = 3	n = 4
Ph•	2025 (40) 2053 (40) 2086 (5)	2038 (101) 2061 (66)	2041 (101) 2061 (168)	2035 (82) 2061 (224)
CH ₂ :C Me·	[≭] 2013 (36) 2035 (37)	2021 (53)	2017 (49) 2039 (58)	2008 (5) 2017 (25) 2039 (113)
deleur ef est	2063 (15) 2089 (2.5)	2064 (33) 2089 (7.5)	2062 (51) 2089 (5)	2062 (80)

	a state when the state of the	
- A & L	A 1-11 1-1	
	ABLE.	
	M 1 1 1 1 1 1	- L
_	the second second second second	

They are useful as a measure of the relative strengths of the various absorptions, but since the solutions contain mixtures of more than one complex no particular significance can be attached to them.

It is significant that in the isopropenyl series solutions for which the P:Cu ratios were three and four were quite colourless, and therefore could not contain appreciable amounts of the bright yellow 1:1 complex.

The acetylenic frequencies in the complexes of phenylethynylgold are collected in Table 7. There is no apparent relation between the frequency and the nature of the ligand and it is remarkable that the frequencies fall within such a narrow range 2107 - 2134 cm⁻¹, (with the exception of that of PhC:CAuPMe₃ at 2095 cm⁻¹), close to those of phenylethynyl-nickel^{75b}, -palladium^{75c} and -platinum(II)^{75a} complexes with triethylphosphine (2100 - 2108 cm⁻¹).

These frequencies are displaced very little from that of bisphenylethynylmercury(II) (2149 cm⁻¹) and π bonding between the metal and the acetylene is therefore not significantly larger.

The acetylene frequencies in the t-butylethynylgold complexes prepared are peculiar since three (and in one case four) weak bands are formed.

TABLE 7

Acetylenic stretching frequencies in complexes PhC:CAuL

L	λ (C:C) cm ⁻¹	${f L}$	λ (C:C) cm ⁻¹
NH ₃	2120 ^a 2121 ^e	PMe ₃	2095 ^a
Pr ⁱ NH ₂	2122 ^a	PEt ₃	2109 ^a
n-C ₅ H ₁₁ NH ₂	2126 ^a 2127 ^b	PPh ₃	2123 ^a <u>vw</u> 2129 ^b <u>vw</u>
n-C8H17NH2	2122 ^a	¹ / ₂ (Ph ₂ PCH ₂) ₂	2125 ^d
$n-C_9H_{19}NH_2$	2125 ^a	AsEt 3	2113 ^a 2128 ^b
Bu2 ⁿ NH	2131 ^a	SbEt ₃	2107 ^a
piperidine	2130 ^a	P(OPh)3	absent
pyridine	2134 ^a		

 $\begin{array}{ccc} & & & & & \\ & & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\ & & & \\$

^a KBr disc, ^b benzene, ^c carbon tetrachloride, ^d Nujol mull, ^e hexachlorobutadiene.

TABLE 8

Acetylenic stretching frequencies in Bu^tC:CAu and its complexes

 $\mathcal{J}(C:C) \text{ cm.}^{-1}$ (KBr)

(Bu ^t C:CAu) ₄	2027 (<u>w</u>)	2002 (w)	1982 (<u>w</u>)
(Bu ^t C:CAu) _x	2000 (sh)	1972 (sh)	1956 (<u>w</u>)
Bu ^t C:CAUNH ₃	2110 (w)	2004 (<u>w</u>)	1968 (<u>vw</u>)
Bu ^t C:CAuPMe ₃	2150 (<u>vw</u>)	2122 (<u>m</u>) 2041 (<u>vw</u>)	1980 (<u>vw</u>)

The frequencies at 2110 and 2122 are similar to those in the corresponding phenylethynylgold complexes and it is strange that the other frequencies are similar to those in the uncomplexed t-butylethynylgold.

Isocyanide frequencies in the gold complexes

The stretching frequency of the isocyanide groups (Table 7) bonded to phenylethynylgold are about 100 cm^{-1} greater than those of the free isocyanides, the differences (bonded) -

(free) being 10, 98 and 81 cm.⁻¹ for the n-butyl-, p-tolyl-, and o-ethylphenyl- isocyanides respectively. Recently Cotton (40-70cm.⁻¹ in Some isocyanide complexes of) and Zingales¹⁴⁸ have reported increases of silver(I) and manganese(II). They find that this effect increases with increase in positive charge on the metal to which the isocyanide is coordinated, that is to say an increase as back co-ordination from the metal decreases. In the gold, complexes the increase in the frequencies are much larger than those described by Cotton and Zingales and the gold atom carries a fractional negative charge. It seems likely, therefore, the increase in $\mathcal{J}(N:C)$ on co-ordination is due to a rise in the bond order of the isocyanide from less than three in the free isocyanide to three in the co-ordinated isocyanide. Thefrequencies of normal cyanide groups (2220 - 2260 cm.⁻¹) are in fact similar to those observed for the isocyanides co-ordinated to the gold atom. N - H frequencies in the gold-amine complexes

The N - H frequencies (Table 9) observed in the phenylethynylgoldamine complexes were lower than those found in the free amines, by some 100 cm⁻¹.

TABLE 9

N - H Stretching frequencies in the complexes PhC:CAu a	mine
amine $N - H^{\frac{\pi}{2}}$ $N - H^{\frac{\pi}{2}}$ in complex in free amin	t # 1e
NH_3 3328 3247 cm ⁻¹	
$n-c_5H_{11}NH_2$ ^a 3287 3232 3380 33	301
$n-C_8H_{17}NH_2$ 3283 3226 3385 33	306
$n-C_{9}H_{19}NH_{2}$ 3264 3224 3377 33	306
$n-(C_4H_9)_2NH$ 3167 3298	
piperidine 3179 3278	

X All KBr discs, except ^a in hexachlorobutadiene XX Taken as liquid films.

This decrease on co-ordination is similar to that of

100-150 cm.⁻¹ found in some platinum complexes¹⁴⁹ and some complexes of other metals.¹⁵⁰ The interpretation of the spectra of amine complexes is frequently complicated by hydrogen bonding (for example, the spectrum of $PPr_3^{n} \cdot NH_3 \cdot PtCl_2$ has six N -H stretching bands even in carbon tetrachloride¹⁴⁹), however, the spectra of the gold-amine complexes are much simpler and little hydrogen bonding seems to occur even in the solid. The bands in the primary and secondary amine complexes are relatively sharp, as sharp as those due to C -H vibrations, and even in the ammonia complexes in which hydrogen bonding effects would be greatest the bands were only slightly broader than the C - H stretching frequencies.

Both the ammonia complexes studied had bands at 1241 cm.⁻¹ (<u>m</u>) in PhC:CAuNH₃ and at 1253 cm.⁻¹ (<u>s</u>) in Bu^tC:CAuNH₃ which were absent in the uncomplexed ethynyl and which can be assigned to a rocking mode. The t-butyl complex showed N - H deformation frequencies at 1623 and 1613 cm.⁻¹ which were obscured in the phenyl complex by bands due to the presence of the phenyl group. Metal-carbon frequencies

Absorptions, which were generally weak, occurred in the copper and silver ethynyls RC:CCu(Ag) in the regions 514-520 and 510-530 cm.⁻¹ respectively, and in the gold ethynyls (PhC:CAu)_x and (Bu^tC:CAu)_x at 526 and 524 cm.⁻¹ respectively. In the t-butyl series (Bu^tC:CCu)₈, (Bu^tC:CAg)_x and (Bu^tC:CAu)₄ similar absorptions were at a slightly higher frequency than in

and at 539, 526, and 560 cm^{3} , respectively, in comparison with (the other ethynyls. These frequencies are attributed to a metal-ethynyl stretching vibration since they are found in the different ethynyls, and by comparison with some previously reported Pt-C¹⁵¹ and Hg-C¹⁵² stretching frequencies and with that of Au-Me stretching at 532 cm.⁻¹ in Et₃PAuMe⁵⁷. On a mass effect along one would expect the frequencies to differ in the sense Cu > Ag > Au, but the observed order, and that of the t-butyl series in particular, suggests that the goldcarbon bonds are stronger than the copper-carbon and silvercarbon bonds.

The ethynyl metal(I) complexes absorb within the narrow range 520-530 cm.⁻¹ in nearly all cases, and these absorptions, also attributed to metal-ethynyl stretching, are weak particularly in the gold complexes. Table 10, collects some of these frequencies which were well-defined, omitting in particular complexes with triphenylphosphite and triphenylphosphine in which the required absorptions are obscured by ligand bands.

TABLE 10

(a)	Metal-ethynyl	stretching	frequencies in	n RC:C-M
	$\mathbb{M} \longrightarrow$	Cu	Ag	Hg
R	Me	absent	absen	t absent
\downarrow	Et	510 <u>vw</u>	512 <u>w</u>	531 <u>m</u>
	Pr	514 <u>w</u>	510 <u>w</u>	544 <u>m</u>
	But	539 <u>m</u>	526 <u>m</u>	542 <u>m</u> , 464 <u>w</u>
	Ph 521 <u>ms</u>	512 <u>ms</u>	530 <u>m</u> 510	0 <u>w</u> 528 <u>m</u>

Table 10 continued

Brc ₆ H ₄		518 <u>m</u>	519 <u>m</u>
NO2C6H4	509 <u>w</u>	516 <u>w</u>	515 <u>w</u>
mesityl	absent	absent	absent

(b) <u>Metal-ethynyl stretching frequencies in RC:C-M-L</u>

 $\partial(c - c_u)$

PhC:CCuPMe3		524		
PhC:CCuPEt3		522 <u>w</u>		
Bu ^t C:CCuPMe3		529 <u>w</u>		
CH ₂ :CMeC:CCu	PMe ₃	541 <u>m</u> , 519) <u>w</u>	
PhC:C.C:CCuP	R' 3	512 - 515	$\underline{\mathbf{W}}$ (R' = Me,	Et, Pr ⁿ)
(PhC:C·C:CCu) ₂ (PEt ₃) ₃	513 <u>w</u>		
	ა(c -	Ag)		
PhC: CAgPMe3		521 <u>w</u>		
	λ(c -	Au)		
Bu ^t C:CAuNH ₃		533 <u>mw</u>		
Bu ^t C:CAuPMe ₃		538 <u>mw</u>	and a start of the	
	PhC:CAu	<u>- L</u>		
L)(C - Au)	\mathbf{T}	$\Im(c - Au)$	
		pyridine	526	
n-C5HllNH2	526	PEt3	526	
$n-C_8H_{17}NH_2$	524	AsEt 3	520	
^{n-C} 9 ^H 19 ^{NH} 2	526			

Table 10 continued

(n-C ₄ H ₉) ₂ NH	529	Bu ⁿ NC	526
piperidine	524	p-tolylNC	526

APPENDIX I

STABLE TERTIARY PHOSPHINE COMPLEXES OF TRIMETHYLGOLD

The only co-ordination stabilized derivatives of trimethylgold, which itself decomposes above -40° , previously described are various amine complexes,¹⁵³ and a thiol complex Me₂AuSPh, which is probably dimeric. The preparation of one of the most stable of these bistrimethyl(1:2 diaminoethane)gold, Me₃AuNH₂CH₂CH₂H₂NAuMe₃ was repeated to ensure that the technique of preparing trimethylgold (from methyl-lithium and gold bromide at -65°) had been acquired, and it has been found to be a suitable starting material for preparing complexes with stronger donors. Solutions in ether, of this complex, rapidly turn blue at room temperature.

Addition of triphenylphosphine to trimethylgold at -65° afforded the colourless crystalline stable complex, Me₃AuPPh₃, which is the first organo-gold(III) phosphine complex to be described. It melts at 120° with evolution of ethane. Further decomposition studies have shown that on heating: Me₃AuPPh₃ \longrightarrow C₂H₆ + MeAuPPh₃ $\longrightarrow \frac{1}{2}C_2H_6$ + Au + PPh₃ This mode of decomposition takes place in solution and it is remarkable that no gases other than ethane are formed. In this respect it resembles the decomposition of methylsilver to ethane and silver as the only products, even when the decomposition is carried out in alcohol.⁵² The yields of ethane were:

solid Me ₃ AuPPh ₃	$\xrightarrow{120^{\circ}}$	87.2%	C2H6
Me ₃ AuPPh ₃ in xylene	<u>120</u>	56.8%	°₂ ^н 6
solid MeAuPPh3	175	95%	^C 2 ^H 6
MeAuPPh3 in diphenyl	<u> 175 </u>	48%	°₂ ^н 6

The complex trimethyl(trimethylphosphine)gold, m.p. 23° , is formed when trimethylphosphine is added to the ethylenediamine complex in ether, at -20° . It can be purified by sublimation at room temperature (0.004 mm pressure) on to a cold finger at -78° .

It now seems likely that other complexes (e.g. with isocyanides and possibly arsines), of trimethylgold may be prepared.

Infra-red spectra of these gold(III) complexes show two absorptions which are assigned to C-Au asymmetrical and symmetrical stretching at 504 and 477 in the ethylenediamine complex, and at 541 and 498 cm.⁻¹ in the trimethylphosphine complex. This region is more complicated in the triphenylphosphine complex due to ligand bands, but similar assignments are made to the absorptions at 530 and 512 cm.⁻¹. It is apparent that C-Au bonding is stronger in the phosphine complexes than in the ethylenediamine complex.

Experimental

Anhydrous goldmbromide

Prepared by adding excess bromine to dry gold powder as described by Gibson and Colles¹⁵⁴, but omitting the glass beads. After removal of excess bromine, the crystalline product is stored over caustic soda, and handled under nitrogen. It is only very slightly soluble in ether.

Trimethyl(l:2 diaminoethane)gold¹⁵³ Me₃AuNH₂CH₂CH₂H₂NAuMe₃

Trimethylgold was prepared from gold bromide (11.5g., 0.026 mole) and methyl-lithium at -65°. Ethylenediamine (1 cc., excess) was added and after stirring for one hour, the solution was allowed to warm up. Addition of water (50cc.) followed by removal of the ether from the organic layer gave a white solid which was recrystallised from ether-petroleum (40-60°) mixture. Yield 4.1g., 57%, m.p. 94° (explodes). <u>Trimethyl(trimethylphosphine)gold Me₃PAuMe₃</u>

Trimethylphosphine (Excess of a standard solution in ether) was added to the ethylenediamine complex (0.5g.) in ether (locc.) at -20° . After adding water (locc.) the ether layer was separated and the ether removed to give oily droplets which solidified on cooling below 0° . Sublimation of this oil at 0.004 mm/23° on to a cold finger at -78° afforded colourless Me₃AuPMe₃, m.p. 23°. (Found: C, ; H, Au, 61.6. C₆H₁₈AuP requires C, ; H, Au, 61.9%).

Trimethyl(triphenylphosphine)gold Me3AuPPh3

Triphenylphosphine (6.8g., 0.026 mole) in ether (20cc.) was added to trimethylgold (from gold bromide ll.3g., 0.26 mole) in ether (150 cc.) at -65° . After stirring for 45 minutes the solution was allowed to warm up and water (50cc.) added. Gold (2.4g.) was filtered off and the ether layer, which contained triphenylphosphine and the required complex, separated. This mixture was separated by elution through an alumina column, the triphenylphosphine remaining on the column. The solid from the ether eluent was crystallised from hexane affording Me_AuPPh_3, 5.3g., 41% m.p. 120° (decomp). (Found: and $50 \cdot 2$ C, 51.1; H, 4.7; Au, 39.5. $C_{21}H_{24}AuP$ requires C, 50.0; H, 4.8; Au 39.1%).

Thermal Decomposition of Me AuPPh 3

On heating trimethyl(triphenylphosphine)gold (0.3516g.) in vacuo, at 120° for half an hour ethane (20.64 N cc., corresponding to 87% of the total available methyl groups', identified by I. R. spectra), gold, and triphenylphosphine (m.p. 79°) were obtained.

Decomposition of a further 0.1646g., in xylene for half an hour at 120° afforded 4.15 N cc. ethane corresponding to 56.8% available methyl groups. A qualitative decomposition in xylene at 120° , gave methyl(triphenylphosphine)gold after removal of the xylene, identified by m.p. 175° (decomp.) and I.R. spectra.

Thermal decomposition of MeAuPPh2

Decomposition of methyl(triphenylphosphine)gold (0.2700g.) at 175° for half an hour gave ethane (5.79 N cc. corresponding to 94.8% available methyl groups), gold and triphenylphosphine (m.p. 79°). Decomposition of a further 0.4568g., in diphenyl, at 175° for half an hour gave 5.49 N cc. ethane, corresponding to 48.3% available methyl groups.

APPENDIX II

THE FAR INFRA-RED SPECTRA OF SOME GOLD COMPOUNDS

It is only recently that spectrometers have become available for investigation of the infra-red region between 20 and 50µ. One such instrument, a Grubb Parsons double beam grating spectrometer has been installed in this department, and assignments have been made to absorption bands present in the spectra of some gold compounds.

The spectra of the compounds listed in Table 11 (page 141) are fairly simple in the region $22-50\mu$, there being generally no more than two or three bands.

Gold-phosphorus stretching frequencies

Comparison of the spectra of some gold-phosphine complexes with each other, and with some other tertiary phosphine complexes has shown that bands which are attributed to \Im Au-P occur in the range 347 - 391 cm.⁻¹ In the complexes, $(Me_3P)_2NiCl_2$, $(Me_3P)_2NiBr_2$ and $(Me_3P)_2NiI_2$, frequencies attributed to \Im Ni-P occur at 361 w, 360w, and 366s respectively.

In the series Me_3PAuX (X = Cl, Br, I) (Table II), there is a steady decrease in \Im Au-P as the atomic weight of the halogen increases, that is to say as the double bond character of Au-X increases there is a corresponding decrease in the gold-phosphorus bond order. In comparison with these frequencies, that at 357 <u>m</u> in Me₃PAuMe is at a much lower wave number than would be expected.

Gold-halogen frequencies

Some transition metal- chlorine stretching frequencies have already been reported¹⁵⁵ for the complex halides of rhenium, palladium, osmium iridium and platinum to be in the range 313-358 cm.⁻¹, and in some of these complexes isotopic splitting of the metal-chlorine bands are observed. The monochlorogold compounds in the table show this isotopic splitting particularly well, and two frequencies due to $\Im Au$ -Cl and $\Im Au$ -Cl are present in these complexes. The $\Im Au$ -Cl frequencies are generally in the range 311-329 cm⁻¹. In the complexes Me₂SAuCl, Me₃AsAuCl, Me₃PAuCl, the $\Im Au$ -Cl frequency decreases as the π -bonding strength of the donor atom (P>As>S) increases, but there is no similar regular change in the analagous bromo- series.

Gold- bromine stretching frequencies occur in the region 210-233 cm⁻¹; gold-iodine frequencies lie outside the range of the spectrometer.

Other assignments in the 22-50µ region

In the trimethylphosphine complexes, in the table, the only other band besides the metal- ligand and metalhalogen absorptions, is one occurring in the region 270-300 cm.¹ and which, by analogy with a similar band in Me_3P0^{156} at 256 cm.¹ and in $Me_3PCl_2^{157}$ at 302 cm.¹, is attributed to a PC3 bending vibration.

In the triphenylphosphine complexes a weak absorption at 444 in Ph_3PAuCl and 441 cm⁻¹ in Ph_3PAuBr must be due to a vibration of the triphenylphosphine group.

TABLE 11

INFRA-RED Au	- P, Au - Cl	and Au - Br STRET	CHING
	FREQUENC	<u>LES</u>	
Compound	Au-P	Au-Cl	Au-Br
Me ₃ PAuCl	381 <u>m</u>	311 <u>s</u> , 303 <u>sh</u>	
Me ₃ PAuBr	378 <u>s</u>		224 <u>s</u>
Me ₃ PAuI	371 <u>s</u>		
Et 3PAuCl		312 <u>s</u> , 305 <u>sh</u>	
Et ₃ PAuBr			210 <u>s</u>
Ph3PAuCl		329 <u>s</u> , 32 <u>3ms</u>	
Ph ₃ PAuBr			233 <u>s</u> , 229 <u>sh</u>
Me ₃ PAuC:CPh	347 <u>s</u>		
Me 3 PAuC: CBut	368 <u>m</u>		
Me ₃ PAuMe ₃	391 <u>w</u>		
Me ₃ PAuMe	357 <u>m</u>		
Et ₃ PAuMe	388 <u>vw</u>		
	<u>→Au-S</u>		
Me ₂ SAuCl	344 <u>m</u>	324.5 <u>s</u> 319.5 <u>sh</u>	
Me ₂ SAuBr			228 <u>m</u>

Table || continued

	2 Au-As		
Me ₃ AsAuCl	265 <u>s</u>	317 <u>s</u> 312 <u>s</u>	
Me ₃ AsAuBr	absent		210 <u>s</u>
Me ₃ AsAuI	268 <u>s</u>		

Experimental

Samples of the compounds were ground in nujol and the mulls mounted between cesium iodide discs. In the case of halides other than iodides, halogen exchange with the cesium iodide was overcome by using thin polythene sheets between the mulls and the discs.

REFERENCES

1.	D. Blake, G. Calvin and G. E. Coates, Proc. Chem. Soc., 1959, 306.
2.	S. J. Ahrland, J. Chatt et. al., J., 1958, 264; 276.
3.	L. E. Orgel, J. 1958, 4186; Introduction to Transition Metal Chemistry Ligand Field Theory, Methuen, London, 1960.
4.	R. S. Nyholm, Proc. Chem. Soc., 1961, 273-296.
5.	 H. Remy and C. Laves, Ber., 1933, <u>66</u>, 571; G. S. Forbes and H. I. Cole, J. Amer. Chem. Soc., 1921, <u>43</u>, 2492; W. Erber and A. Schiihly, J. prackt. Chem. 1941, <u>158</u>, 176.
6.	W. Biltz and W. Stollenwerk, Z. anorg. chem., 1920, 114, 174.
7.	H. H. Morgan, J., 1923, <u>123</u> , 2901
8.	P. Dutoit and L. Friderich, Bull. Soc. chim. France, 1898, [3], <u>19</u> , 321.
9.	F. G. Mann, A. F. Wells and D. Purdie, J., 1937, 1828.
10.	G. T. Morgan and F. H. Burstall, J., 1928, 143.
11.	R. Adams and A. Ferretti, J. Amer. Chem. Soc., 1959, 81, 4927
12.	H. Rheinboldt, F. Molt and E. Motzkus, J. prackt. Chem., 1932, [2], 134, 257; R. Adams et. al., Croatica Chem. Acta., 1957, 29, 277-285; Chem. Abs., 1959, 43, 16145.
13.	A. Rosenheim and S. Sleinhäuser, Z. anorg. chem., 1900, <u>25</u> , 72.
14.	G. T. Morgan and W. Ledbury, J., 1922, <u>121</u> , 2882.
15.	R. Holtje and H. Schlegch, Z. anorg. chem., 1939, <u>243</u> , 246; R. Scholder and K. Pattack, ibid., 1934, <u>220</u> , 250.
16.	F. G. Mann, D. Purdie and A. F. Wells, J., 1936, 1503; F. G. Mann and A. F. Wells, Nature, 1937, <u>140</u> , 502.
17.	A. F. Wells, Z. Krist., 1937, <u>94</u> , 447.
18.	R. C. Cass, G. E. Coates and R. C. Hayter, Chem. and Ind., 1954, 1485.
19.	F. A. Cotton and D. M. L. Goodgame, J. 1960, 5267.

	20.	R.	S. Nyholm, ibid., 1952, 1257.
	21.	C.	M. Harris, R. S. Nyholm and N. A. Stephenson, Rec. Trav. Chim., 1956, <u>75</u> , 687
	22.	₩.	Cochran, F. A. Hart and F. G. Mann, J., 1957, 2816.
	23.	Α.	E. Arbuzov and V. M. Zoroastrova, Chem. Abs. 1953, <u>47</u> , 9898.
	24.	M.	Levi-Malvano Atti. R. 1908, <u>17</u> , 857.
	25.	L.	Lindet, Chem. Rev., 1884, <u>98</u> , 1282.
	26.	J.	V. Berkade and T. S. Piper, Advances in the Chemistry of the Co-ordination Compounds, Macmillan, New York, 1961.
	26 ^a .	id	em., unpublished observations.
	27.	H.	Grossmann and P. V. D. Forst, Z. anorg.chem., 1905, <u>43</u> , 94; J. Brigando, Chem. Rev., 1942, <u>214</u> , 908.
	28.	D.	T. Cromer, J. Phys. Chem., 1957, <u>61</u> , 1388.
	29.	J.	L. Hoard, Z. Krist., 1933, <u>84</u> , 231; D. T. Cromer unpublished work.
	30.	id	em., J., 1939, 426.
	31.	Г.	Malatesta, Progress in Inorganic Chemistry, Vol. I, p. 283-379, Ed. F. A. Cotton, Interscience, New York.
	32.	H.	Guillemard, Ann. Chim. Phys., 1908, [viii], 14, 344.
	33.	E.	G. J. Hartley, J., 1928, 780.
	34.	F.	Klages, K. Monkemeyer and R. Heinle, Ber., 1952, 85, 109.
	35•	Ε.	G. J. Hartley, J., 1916, 1296.
	36.	K.	A. Hoffmann and G. Bugge, Ber. 1907, <u>40</u> ,1772.
	37.	A.	Sacco and M. Freni, Gozz. chim. ital., 1956, 85, 195.
	38.	Ρ.	L. Robinson and K. R. Stainthorpe, Nature, 1944, 153, 24
. '	39.	H.	Bloom, Nature, 1947, 159, 539; Chem. Abs., 1944, <u>38</u> , 4875.
4	40.	₩.	A. Jones, J. Amer. Chem. Soc., 1899, 22, 287; W. Manchot and J. A. N. Friend, Ann., 1808, 359, 100; O. H. Wagner, Z. anorg. chem., 1931, 196, 364.

41.	W.	Manchot and H. Gall, Ber., 1924, <u>57</u> , 1157.
42.	R.	Nast and C. Schultze, Z. anorg. chem., 1960, 307, 15.
43.	₩.	Manchot and H. Gall, Ber., 1925, <u>58</u> , 2175
44.	Μ.	S. Kharasch and H. S. Isbell, J. Amer. Chem. Soc. 1930, 52, 2919
45.	re	ference 1, page 284
46.	Η.	Gilman, R. G. Jones, and L. A. Woods, J. Org. Chem. 1952, <u>17</u> , 1630.
47.	D.	T. Hurd and E. G. Rochow, J. Amer. Chem. Soc., 1945, <u>67</u> , 1057.
48.	H.	Gilman and J. M. S. Straley, Rec. Trav. Chim., 1936, <u>55</u> , 821.
49.	C.	E. H. Bawn and R. Johnson, J., 1960, 3923; 3296.
50.	F.	A. Bolth, W. M. Whaley and E. B. Starkey, J. Amer. Chem. Soc., 1943, <u>65</u> , 1456
51.	G.	Wilkinson and T. S. Piper, J. Inorg. Nuclear Chem. 1956, 2, 32.
52.	c.	E. G. Bawn and F. J. Whitby, Discuss. Faraday Soc., 1947, <u>2</u> , 228; H. Theile, Z. Electrochem., 1943, <u>49</u> , 426; G. Semerano and L. Riccoboni, Ber. 1941, <u>74</u> , 1089.
53.	F.	Glockling, J., 1955, 716; 1956, 3640
54.	R.	Reich, Compt. rend., 1923, <u>177</u> , 322; E. Krause and B. Wendt, Ber., 1923, <u>56</u> , 2064.
55.	F.	Challenger and C. F. Allpress, J., 1921, 916.
56.	F.	Glockling and M. Jewitt, unpublished observations.
57.	G.	Calvin, G. E. Coates and P. S. Dixon, Chem. and Ind., 1959, 1628.
58.	G.	E. Coates, Organo-Metallic Compounds, Second Edition, Menthen, London, 1960.
59.	Ν.	V. Sidgwick, Chemical Elements and Their Compounds Vol. I., p.181-188, Oxford, 1950.
60.	М.	Berthelot, Ann. Chim. Phys., 1901, 23, 32.

- 61. H. Tropsch and W. J. Mattox, J. Amer. Chem. Soc., 1935, 57, 1102; Inorg. Synth., Vol. VI, 217.
- 62. H. B. Jonassen and J. E. Field, J. Amer. Chem. Soc., 1957, 79, 1275.
- 63. R. E. Kepner and L. J. Andrews, J. Org. Chem., 1948, 13, 208; J. Amer. Chem. Soc., 1949, 71, 1723; R. M. Keefer, L. J. Andrews and R. E. Kepner, ibid., 2381, 3906; L. J. Andrews and R. M. Keefer, ibid., 1948, 70, 3261.
- 64. W. F. Eberg, H. L. Welge, D. M. Yest and H. J. Lucas, J. Amer. Chem. Soc., 1937, <u>59</u>, 45; S. Winstein and H. J. Lucas, ibid., 1938, <u>60</u>, 836; F. R. Hepner, K. N. Treblood and H. J. Lucas, ibid., 1952, <u>74</u>, 1333, 1338
- 65. A. C. Cope and L. L. Esters, ibid., 1950, <u>72</u>, 1128; A. C. Cope, C. L. Stevens and F. A. Hochstein, ibid., 2510; W. O. Jones, J., 1954, 312.
- 66. H. J. Lucas, R. S. Moore and D. Pressman, J. Amer. Chem. Soc., 1943, <u>65</u>, 227; H. S. Taufen, M. J. Murray and F. F. Cleveland, ibid., 1941, <u>63</u>, 3500.
- 67. M. J. S. Dewar, Bull. Soc. Chim., 1951, 18, c, 79
- 68. J. Chatt and L. A. Duncanson, J., 1953, 2939.
- 69. A. E. Comyns and H. J. Lucas, J. Amer. Chem. Soc., 1957, 79, 4339, 4341
- 70. Organo-metallic chemistry, Chapter 10; Editor, H. Zeiss, Chapman and Hall, New York 1960; International Conference on Co-ordination Chemistry, Chem. Soc. Special Publication No. 13, London 1959.
- 71. J. Chatt, L. A. Duncanson and R. G. Guy, Chem. and Ind., 1959, 430; J., 1953, 2939.
- 72. F. L. Carter and E. W. Hughes, Acta. Cryst. 1957, 10, 801
- 73. F. L. Carter and H. J. Lucas, J. Amer. Chem. Soc., 1957, <u>79</u>, 1306
- 74. J. Chatt, G. A. Rowe and A. A. Williams, Proc. Chem. Soc., 1957, 208.
- 75. (a) J. Chatt and B. L. Shaw, J., 1959, 705, 4020; (b) idem., ibid., 1960, 1718; (c) G. E. Coates and G. Calvin, J., 1960, 2008.

76.	R.	Nast and W. Pfab, Ber., 1956, <u>89</u> , 415
77.	id	em., Z. anorg. chem., 1957, 292.
78.	R.	Nast and E. Sirtl, Ber., 1955, <u>88</u> , 1723
79.	R,	Nast and H. Griesshammer, Ber., 1957, <u>90</u> , 1315
80.	R.	Nast and F. Urban, Z. anorg. chem. 1956, 287, 17
81.	R.	Nast and H. Lewinsky, Z. anorg. chem. 1955, 282, 210
82.	R.	Nast and H. Kasperl, Ber., 1959, <u>92</u> , 2135
83.	R.	Nast and K. Vester, Z. anorg. chem., 1955, 279, 146.
84.	R.	Vestin and I. Ralf, Acta. Chem. Scand., 1949, 3, 101
85.	Α.	J. Hill and F. Tyson, J. Amer. Chem. Soc., 1928, <u>50</u> , 172
86.	C.	A. Young, R. R. Vogt and J. A. Nieuland, J. Amer. Chem. Soc., 1936, <u>58</u> , 55.
87.	R.	A. Raphael, Acetylenic compounds in Organic Syntheses, Butterworths, London 1955, p.19
88.	N.	A. Milas, O. L. Mageli, J. Amer. Chem. Soc., 1953, <u>75</u> , 5970.
89.	F.	Sondheimer, Y. Amiel and R. Wolovsky, J. Amer. Chem. Soc., 1957, <u>79</u> , 4247; ibid., 1959, <u>81</u> , 4600; G. Eglinton and A. R. Galbraith, J., 1959, 889.
90.	Α.	L. Klebanskii, I. V. Grachev and O. M. Kuznetsova, Zhw. Obschei. Khim. 1957, <u>27</u> 2977; Chem. Abs., 1958, <u>52</u> , 8034
91.	G.	Calvin, Ph.D. Thesis, Durham, August 1959.
92.	D.	Blake, Ph.D. Thesis, Durham, May 1959.
93.	J.	A. Mathews and L. L. Waters, J. Amer. Chem. Soc., 1900, <u>22</u> , 108.
94.	0.	Y. Okhlobystin and L. I. Zakharkin, Izvest. Akad. Nauk. S.S.S.R; Otdel. Khim. Nauk., 1958, (8), 1006
95.	Μ.	Sander, Angew. Chem., 1961, <u>73</u> , 67
96.	K.	S. Pitzer and H. S. Gutowski, J. Amer. Chem. Soc., 1946, 68, 2204.

910	A.	v. Grosse and J. M. Mavruy, J. Urg. Onem., 1940, $\underline{2}$, 100
98.	F.	G. Mann and A. F. Wells, J., 1938, 708.
99.	C.	M. Kosolapoff, Organo Phosphorus Compounds, Wiley, New York, 1950, p.184.
100.	R.	A. Raphael, Acetylenic Compounds in Organic Syntheses, Butterworths, London 1955, p.19.
101.	J.	H. Allan, E. R. H. Jones and M. C. Whiting, J., 1955, <u>77</u> 1870.
102.	· C .	R. Noller and R. Adams, J. Amer. Chem. Soc., 1924, <u>46</u> , 1893.
103.	0r	ganic Reactions, <u>5</u> , 50.
104.	Τ.	H. Vaughan and J. A. Nieuwland, J. Amer. Chem. Soc., 1934, 56, 1207.
105.	Ε.	A. Braude, E. R. H. Jones and E. S. Stern, J., 1947, 1096
106.	J.	B. Armitage et. al., J. 1954, 152.
107.	H.	H. Schlubach and V. Franzen, Annalen, 1951, 573, 105
108.	Ind	org. Synth. 2, 1.
109.	H.	Lund and J. Bjerrum, Ber., 1931, <u>64</u> , 210.
110.	Ρ.	Walden and O. Werner, Z. physick. Chem., 1924, 111, 465
111.	K.	C. Bailey J., 1939, 767.
112.	Α.	Findlay, Practical Physical Chemistry, Longmans, 1949, p.125
113.	Μ.	Faraday, Experimental Researches, 1838, 699.
114.	R.	Clausius, Die Mechanische Wormetheorie, (11), Braunschweig 1879, 62
115.	0.	F. Moscotti, Mem. Math. Fis. Modena, 24, (11), 1850, 49
116.	J.	C. Maxwell, Trestise on Electricity and Magnetism (11), 1881, 396
117.	Т.	Lorenz, Ann. phys. Lpz. <u>11</u> , 1880, 70
118.	H.	A. Lorentz, ibid., <u>9</u> , 1880, 641

119.	P. Debye, Phys. Z., 1912, <u>13</u> 97; P. Debye, Handbuch der Radiologie, 1925, <u>6</u> , 597
120.	D. H. Wiffen, Trans. Faraday Soc., 1958, <u>54</u> , 327
121.	I. F. Halverstadt and W. D. Kumler, J. Amer. Chem. Soc., 1942, <u>64</u> , 2988
122.	P. Dixon, Ph.D. Thesis, Durham, August 1960.
123.	J. G. Livingstone, Ph.D. Thesis, Durham, May, 1961
124.	L. A. Sayce and H. V. A. Briscoe, J., 1925, 315
125.	R. A. W. Hill and L. E. Sutton, J., 1953, 1482
126.	L. Hartshorn and D. A. Oliver, Proc. Roy. Soc., 1929, <u>123A</u> , 683.
127.	Technique of Organic Chemistry, Vol. VII, 2nd Edition, Interscience, New York, 1955, p.422.
128.	Applied Inorganic Analysis, W. F. Hillebrand, et. al. 2nd Edition, Wiley, New York, 1955, p.250.
129.	I. M. Kohltoff and E. B. Sandell, Textbook of Qualitative Inorganic Analysis, 3rd Edition, Macmillan, New York, 1953, p.600.
130.	ref. 128, p.217
131.	ref. 128, p.207
132.	ref. 129, p.538
133.	F. Straus, Annalen, 1905, <u>342</u> , 224
134.	Beilsteins Handbuch der Organische Chemie, Band V, p.514
135.	J. R. Johnson and W. L. McEwan, J. Amer. Chem. Soc., 1926, 48, 469.
136.	T. H. Vaughn, ibid., 1933, <u>55</u> , 3453
137.	S. J. Cristal et. al., ibid., 1954, <u>76</u> , 4558
138.	M. M. Otto, ibid., 1934, <u>56</u> , 1393
139.	M. Nakagawa, J. Chem. Soc. Japan (Pure Chem. Section) 1951, 72, 561.

140.	Ε.	O. Fischer, W. Hafner, and H. O. Stahl, Z. anorg. Chem., 1955, <u>47</u> , 282
141.	Ŀ.	Favorski and A. Morev, J. Russ. Phys. Chem. Soc., 1920, 50, 571
142.	Ρ.	W. R. Corfield and H. M. M. Shearer, personnal communi- cation.
143.	I.	E. Orgel, Introduction to Transition Metal Chemistry, Ligand Field Theory, Methuen, London 1960.
144.	R.	G. Hayter, Nature, 1962, <u>193</u> , 872
145.	L.	J. Bellamy, The Infra-red Spectra of Complex Molecules, 2nd Edition, Menthen, London, 1958.
146.	J.	Chatt, L. A. Duncanson and R. G. Guy, Chem. and Ind., 1959, 430.
147.	J.	Chatt and L. A. Duncanson, J., 1953, 2939
148.	F.	A. Cotton and F. Zingales, J. Amer, Chem. Soc., 1961, <u>83</u> , 351
149.	J.	Chatt, L. A. Duncanson and L. M. Venanzi, J., 1955, 4461; 1956, 2712
150.	J.	V. Quaglinano, C. Curran and G. F. Svatos, J. Amer. Chem. Soc., 1955, <u>77</u> , 6159.
151.	D •	M. Adams, J. Chatt, and B. L. Shaw, J., 1960, 2047.
152.	D.	R. J. Boyd, H. W. Thompson and R. L. Williams, Discuss. Faraday Soc., 1950, <u>9</u> , 154; Gutowsky, J. Chem. Phys., 1949, <u>17</u> , 128
153.	H.	Gilman and L. A. Woods, J. Amer. Chem. Soc., 1948, 70, 550
154.	C.	S. Gibson and W. M. Colles, J., 1931, 2411
155.	D.	M. Adams, Proc. Chem. Soc., 1961. 335.
156.	J.	Goubeau and W. Berger, Z. anorg. Chem., 1960, 304, 147
157.	J.	Goubeau and Baumgärtner, Z. electrochemie, 1960, 64, 598.

A C NICH 1967